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APPLICATION OF THE WAVY MECHANICAL FACE SEAL
TO SUBMARINE SEAL DESIGN

BY

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APPLICATION OF THE WAVY MECHANICAL FACE SEAL
TO SUBMARINE SEAL DESIGN

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20. ABSTRACT (Continue on reverse side if necessary and identify by block number) → This report details two years of work on the development of a long life submarine seal. In the first portion of the report, three different long life designs based on the waviness concept are described. Design details and test results are given. The conclusions reached are that the first design gives long life but is too complicated. The second design is much simpler but still gives long life. The third design is very simple but life may be compromised. Even so, the fixed wave (third design) appears to be the best way to apply the		

waviness principle. A new type of grinding apparatus was designed to produce the wave.

Details of seal ring deflection calculations are given. A ring finite element and ring deflection program are given. A contact model is presented. Deflection studies on actual submarine seal rings are presented, and it is shown how present designs are sensitive to lock ring groove waviness. Experimental verification is provided for the contact model. Work on a nonlinear joint model is described.

Experiments to validate the existence of microasperity lubrication in water are described and dismissed. A unique apparatus was constructed. Experiments suggest microasperities do not cause a large hydrodynamic load support in water.

Several seal analysis programs are described. These programs are very useful for evaluating seal designs. One of the programs provides for a fully automated heat transfer analysis of seal pairs.

Finally, new concepts in squeeze seals and bearings are described. Some numerical results are given showing how in some cases the squeeze seal is the same as at present but in other cases it is different.

A summary and detailed conclusion chapter is given at the end of the report.

APPLICATION OF RESEARCH TO THE NEEDS OF THE U.S. NAVY

Mechanical face seals are used in numerous applications in Naval machinery. These applications range from propeller shaft seals to boiler feed pump seals. In such equipment the mechanical seal plays a vital role. When such seals fail, repair is costly both in terms of lost time and direct costs, so any improvement in seal life and reliability would be of significant benefit.

As more advanced equipment is designed, it is sometimes difficult to achieve desired performance in more severe service environments with the present state of the art of seal design. Thus, an improvement in seal technology would serve this important application.

One objective of the research herein is to further the understanding of mechanical face seal lubrication phenomena. Another objective is to develop the capability of designing contacting face seals having a longer life, greater reliability, and for extreme environments. The immediate objective herein is to apply the knowledge gained to the design of a small scale submarine type seal having exceptionally long life and high reliability.

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LIST OF SYMBOLS

a	Ring cross sectional area - m^2
$A = \frac{EJ_x}{GJ_\theta}$	Stiffness ratio - dimensionless
$B = \frac{r_o^2 - r_b^2}{r_o^2 - r_i^2}$	Balance ratio for an outside pressurized seal
E	Youngs modulus - N/m^2
F	Force - N
$[F]$	Load matrix
G	Shear modulus - N/m^2
h	Nominal film thickness - m
h_n	Amplitude of the nth harmonic - m
I_x, I_y, I_{xy}	Moments of inertia based on straight beam theory
$J_x = \int_A \frac{y^2}{1 - x/r_c} dA$	Moment of inertia about x axis for circular ring cross section - m^4
$J_y = \int_A \frac{x^2}{1 - x/r_c} dA$	Moment of inertia about y axis for circular ring cross section - m^4
$J_{xy} = \int_A \frac{xy}{1 - x/r_c} dA$	Product of inertia for a circular ring
J_θ	Torsional stiffness constant for ring cross section - m^4
k_F, k_S	Spring constants representing face contact and support contact
K	Stiffness matrix
$[K]$	Stiffness matrix for two ring model support contact

$[K_R]$	Individual ring stiffness matrix
m_x, m_y, m_θ	Distributed moments on seal ring $N \cdot m/m$
M_x, M_y, M_θ	Ring moments - $N \cdot m$
n	Number of the harmonic or number of waves around seal face or normal direction
N	Number of nodes
N_θ	Ring normal force - N
p	Fluid pressure - N/m^2
p_c	Cavity pressure - N/m^2
p_i	Seal inside pressure - N/m^2
p_m	Pressure at asperity contact--equals compressive strength - N/m^2
p_o	Seal outside pressure - N/m^2
p_s	Shear strength of asperities - N/m^2
p_{sp}	Spring pressure on face - N/m^2
Q	Total leakage for the seal - m^3/s
r	Radial coordinate
r, θ, z	Seal coordinates
r_b	Seal balance radius - m
r_c, R_c	Radius to centroid of seal ring - m
r_f	Friction radius - m
r_i	Inside radius of seal - m
r_o	Outside radius of seal - m

R	Radius to centroid of seal ring - m
t	time - s
T_q	Seal friction torque - N · m
u, v, w, ϕ	Ring displacements
u, v, w	Displacements in torsion problem
U	Velocity - m/s
v	Ring centroid displacement - m
V_x, V_y	Ring shear forces - N
x, y, θ	Ring coordinates
x, y, z	Coordinate in torsion problem
W	Total load support - N
W^*	Required load support - N
δ	Displacements
$[\delta]$	Displacement matrix
δ_{FO}, δ_{SO}	Initial gap between springs and mating surface for face and support
Δ	Net deflection at the face
η	Vicosity - N · s/m ²
θ	Angular coordinate or flow variable
θ_1, θ_2	Angular coordinates of end nodes of ring element
ψ	Stress functions or angle of twist about y axis
ρ	Density - kg/m ³
Γ^*	Warping constant - m ⁶
ϕ	Rotation of seal ring about its centroid
ω	Angular speed

ν Poisson's ratio

Subscripts

e Refers to one element

P, M Refers of primary and mating rings of seal

F, S Refers to face contact and support contact

1, 2 Refers to nodes at ends of element

CHAPTER 1

INTRODUCTION

Mechanical Face Seals

Applications of face seals range from boiler water feed pumps to compressors, to petrochemical process pumps, to propeller shafts. In many applications the reliability of the mechanical seal is of the greatest importance to the reliability of the equipment itself.

Mechanical face seal technology has been steadily improving over the past several decades. However, there still remain demands for seal performance which have not been met. One such application of note is the submarine propeller shaft.

Such demands on a particular technology can often be satisfied by first improving the technology. In the case of mechanical face seals, the main barrier to advancement has been that the mechanics of seal operation are not well enough understood to be able to reasonably anticipate seal performance as a function design parameters.

During the past eight years of this research program, much has been learned about controlling the hydrostatic and hydrodynamic mechanisms which enhance face seal operation. In this present work, this knowledge is applied to the design of an improved small scale submarine shaft seal. Theory, design, and test results are presented. The results show promise that significant improvements in submarine shaft seal design are possible, and the application of such designs could greatly increase the length of trouble-free shaft seal service.

Seal Lubrication

As background, the mechanical face seal consists basically of two annular rings which rotate relative to each other and which are pressed together by spring and fluid pressures (see Figure 1-1). In conventional seals, the surfaces that rub together are generally manufactured as flat as possible initially so as to minimize leakage. The effective gap between the faces is ideally quite small (order of 1 μm) so that leakage flow across the faces will be quite small. The difficulty in designing a mechanical seal is in maintaining the gap at a very low value while at the same time providing a definite lubricant film between the faces.

The load that must be supported at the faces of a mechanical seal is due primarily to loading caused by the sealed pressure. The load support at the faces is derived from fluid pressure and mechanical pressure. If the fluid pressure at the faces is large enough to support all of the load, then there will be no contact and no adhesive wear.¹ If none of the load is supported by fluid pressure, the load must be carried by mechanical contact, and the wear rate will be large.

In practice, conventional seals often operate at one of two extremes. At one extreme, a large gap will be created by hydrostatic or hydrodynamic pressure or distortion, all of the load will be supported by fluid pressure, and the seal will leak a lot and wear very

¹ There may still be abrasive or corrosive wear even if the surfaces do not touch.

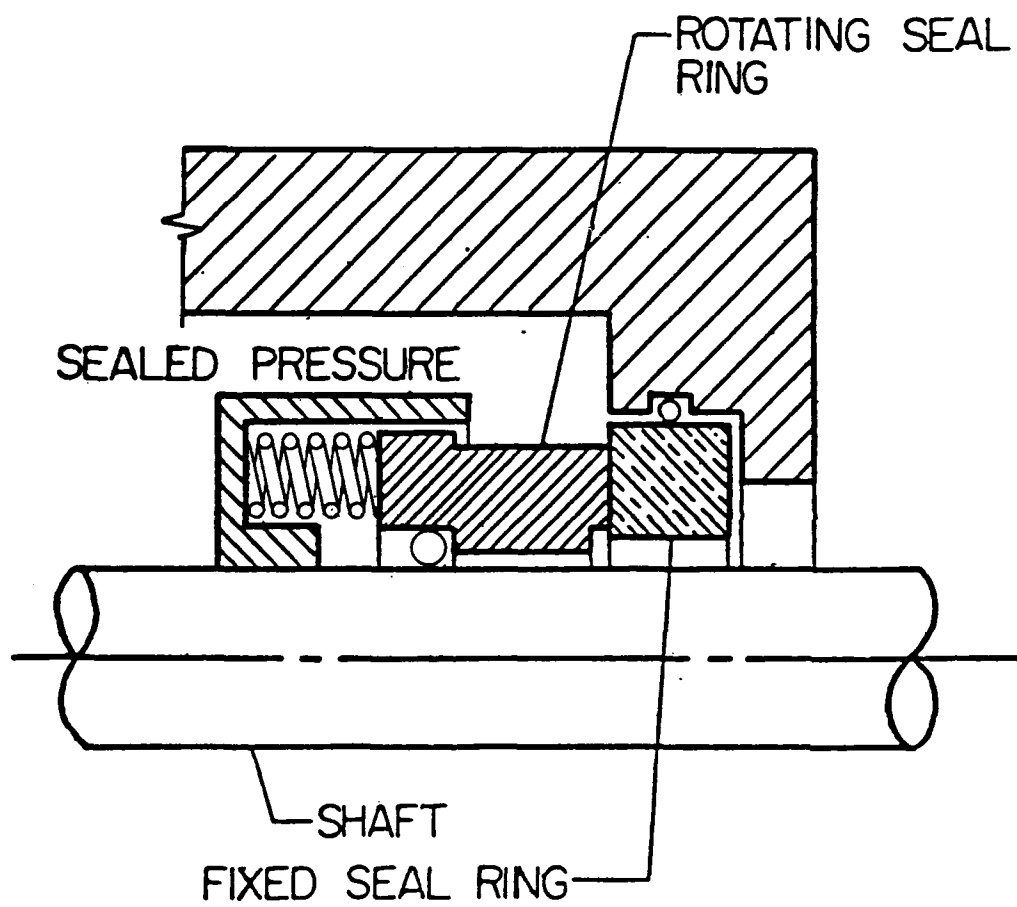


Figure 1-1. Mechanical Face Seal.

little. At the opposite extreme, the gap will close completely. Leakage will be low but only a fraction of the load will be carried by fluid pressure, and wear and heat generation will increase.

Based on the above, it can be concluded that an effective seal should operate between these two extremes--having both adequate fluid pressure load support and low leakage. The seal should operate so that it just touches to minimize leakage but such that the load is carried by fluid pressure. To do this requires that any fluid pressure generation mechanism used to provide load support to the seal must be very carefully controlled. At present in commercial seals, this is left in part to chance and sometimes seals operate at one of the undesirable extremes mentioned.

In this research program, attention has been devoted toward studying the effects of waviness as a source of controlled hydrodynamic and hydrostatic load support. Waviness was selected because it is controllable. In this present work, waviness has been applied as the basis for the design of an improved small scale submarine shaft seal.

Background

ONR-sponsored research on mechanical seals had been conducted for five years prior to the beginning of the submarine seal design phase described herein. The work being reported now evolved from various discoveries over this initial five year period, and the past work must be reviewed to better understand the nature of the current work.

As a starting point for this Navy research program, the effects of waviness on seal performance were modeled in some detail. In the first

annual report for this project, Reference [1], this general problem was solved using a one-dimensional theory. In the second annual report, [2], the much more complex two-dimensional solution to the above problem was solved. The effects of waviness, roughness, asperity contact, wear, cavitation, and elastic deflection were included in this model. Using this model, predictions were made for the relative wear rate, friction, and leakage as a function of roughness, waviness, speed, size, pressure, viscosity, and material.

A number of conclusions were reached based on these first two annual reports:

- 1) The effect of roughness on hydrodynamic lubrication are not completely understood. Certain fundamental questions remain concerning the roughness model used.

- 2) As to the potential of utilizing hydrodynamic effects caused by parallel face waviness to advantage by design, the results show that wear rate and friction can be greatly reduced while maintaining leakage at acceptable levels.

- 3) While a comparison of predicted results to experimental results given in the literature is generally good, data contained in the literature is incomplete, so more complete experimental data are needed for comparison.

- 4) In low viscosity or heavily loaded applications where some roughing is expected to occur, waviness will wear away with time and any benefit derived will be lost unless something is done to counteract this effect.

5) Based upon data for some commercial seals and using the model, it was determined that there was insufficient accidentally caused waviness to produce significant hydrodynamic effects in water. One cannot generalize to say that such effects do not occur in commercial seals. However, using the model the question can be answered on a case by case basis.

Item 1) was treated extensively in the third annual report [3]. Even after this analysis certain fundamental questions remain concerning how to deal with roughness in lubrication problems. However, this thorough analysis led to conclusions allowing certain simplifying assumptions discussed in the fourth annual report [4].

Item 2) was also treated extensively in the third annual report [3]. A methodology for the design of a wavy face seal was developed and applied. Theoretical results showed large reduction in friction and wear rate compared to conventional designs whereas leakage could be controlled.

Concerning Item 3), the second and third annual reports [2,3] describe a test apparatus designed to test the wavy seal theory. This apparatus has been in operation for more than five years and many tests have been conducted. These test results are reported in the fourth and fifth annual reports [4,5].

Early in the test program it was observed that the type of waviness which can be practically applied is not of the radially parallel type. Waviness generally consists of alternating tilt plus radially parallel waviness. Based on these considerations, a new model for

predicting performance was developed and appears in the fourth annual report [4].

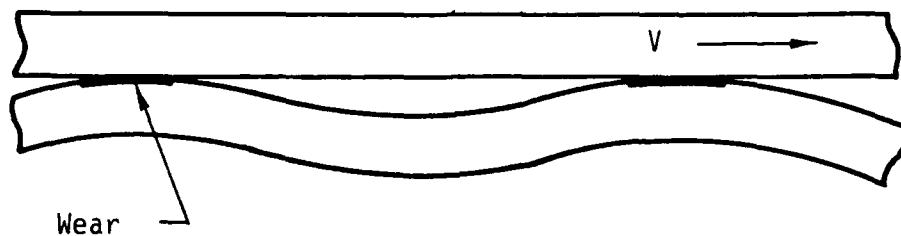
Concerning Item 4) above, a solution to this problem was first proposed in the first annual report [1]. It was proposed to move the waviness slowly around the seal so that whatever wear occurred would be uniformly distributed. Then the shape of the wave would be preserved and tests using a constant wave could be made. The concept is illustrated in Figure 1-2. This concept was incorporated into the test apparatus and is described in detail in References [2] and [3]. In the fifth annual report [5], a new concept to move the waviness with no internal moving parts is described in detail.

In the fifth annual report [5], additional experimental results using waviness are presented, the wavy seal model is further improved, and theory and experiment are compared. The results using a new concept, that of a self-generating seal profile, are reported. Results from tests of the effects of radial taper and high temperature environment are also reported and compared to theory.

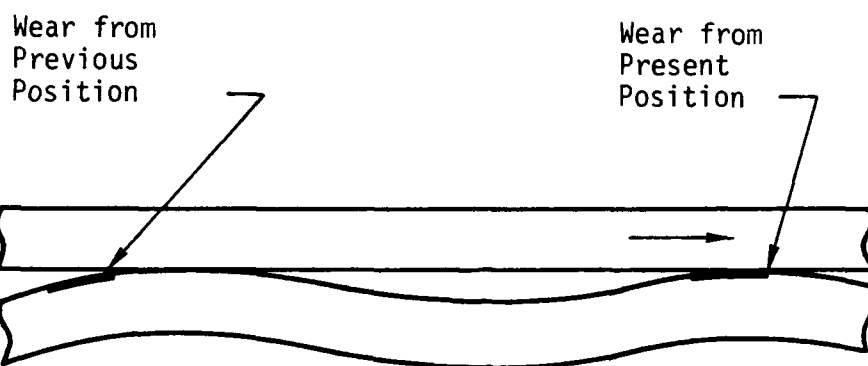
Both the theoretical and experimental basis for applying waviness to a mechanical seal to reduce friction and wear were well established during these first five years.

Submarine Seals

Starting in December 1980, under joint NAVSEA-ONR sponsorship, a three-year submarine seal program was initiated. The general objectives of the three-year effort were:



Distorted Ring



Distorted Ring after Wave Has Moved

Figure 1-2. Moving Waviness Concept.

- 1) to conduct further experiments using moving waviness to further the understanding of this concept.
- 2) to refine mathematical models already developed so as to be able to better predict performance.
- 3) to demonstrate in a practical way the use of the concept of moving waviness to reduce friction and wear in mechanical seals.
- 4) to undertake the development needed to be able to demonstrate the applicability of the concept to submarine shaft seals and to design a full scale long life seal.
- 5) to explore other uses of the waviness concept such as for gas seals and to create better methods by which the concept can be applied.

From these objectives, the following specific tasks were defined:

- 1) Design, fabricate, and test a nine wave optimum seal. Nine waves are needed to minimize changes in tilt with changes in pressure and speed.
- 2) Operate the seal for an extended period of time under simulated submarine operating conditions including seawater and changing speed, pressure, and alignment.
- 3) Evaluate compliancy of existing and proposed submarine seals.
- 4) Conduct friction and wear tests on carbon and hard face materials.
- 5) Seek alternate methods of applying moving waviness.
- 6) Perform analysis of a moving wave gas seal.

In the most recent annual report [6] the progress made on these items is reviewed in detail. Preliminary nine wave test results were reported. Modifications to the test apparatus, design methods and problems and seal deflection calculations were discussed. Details of the progress on all of the above items are contained in this current annual report.

It became clear during the three year period that more work was needed on the development of the wavy seal particularly in the simplification of the waviness mechanism itself. In addition, other submarine seal related problems were identified. Thus this led to a proposal for a fourth year of submarine seal development work, the specific objectives of which were:

- 1) Complete a long term test of a split wavy seal.
- 2) Make a preliminary squeeze seal design.
- 3) Perform a final design and fabricate a new waviness drive mechanism.
- 4) Evaluate seal conformability.
- 5) Evaluate the effect of dirt on a wavy seal.
- 6) Make a face width study for parallel face seals.
- 7) Perform a review of recent submarine seal designs.
- 8) Make a preliminary full scale design.

Considerable progress has been made on these items in addition to the previous list and is reported herein. In particular a wavy seal design which eliminates the need for a waviness drive is proposed and preliminary test results suggest that an ideal wavy seal design may be in hand.

This entire research program spans a number of years. Thus, the entirety of the technical work is reported in six previous annual reports [1-6], this report, sixteen published papers [7-22], and seven master's theses [23-29].

Wavy Face Seal

The concept of waviness is that the film thickness varies in some fashion circumferentially around the seal. Generally speaking, film thickness may vary radially as well as tangentially.

$$h = h(r, \theta) . \quad (1-1)$$

In the present work interest is focused upon film thickness shapes of the following functional form

$$h = h_0 + f(r) \cos n\theta . \quad (1-2)$$

At any particular radius r the film shape is periodic with n waves around the seal and is therefore wavy. However, film shape can also vary in some general manner with r .

If $f(r) = \text{const}$ then the faces are always radially parallel. This component of film thickness variation is commonly termed waviness. If $f(r) \neq \text{const}$, then the faces are not in general radially parallel. $f(r)$ is referred to as tilt. Thus, the film thickness shapes of interest are combinations of waviness and tilt. Since at any radius the film thickness is wavy, the combination of waviness and tilt defined above will also be called a wavy film shape.

The reason for choosing film shapes as described by Equation (1-2) as a subject for study is that these shapes can conveniently be generated by planned mechanical distortions in a seal ring, and the shapes also include common modes of unplanned distortion found in operating seals. For example, generally seals undergo a uniform tilt due to pressure and thermal deformation. When rings are loaded by any non-axisymmetrical load they become wavy as described by Equation (1-2).

Now, for the sake of illustration, assume a seal has a wavy film thickness shape given by Equation (1-2) where $f(r)$ is a piecewise linear function of r . This gives a shape as shown in Figure 1-3 where for each of the n periods a flat mating seal ring will touch all across at the high points and is radially convergent π/n radians away. At any radius, the seal is wavy circumferentially. The waviness enhances hydrodynamic effects and the radial convergence enhances hydrostatic effects. The region where contact is uniform acts as a sealing dam.

It is this shape which previous research has shown has the greatest potential to reduce friction and wear while holding leakage at very low levels. This is the film shape used for the design of the small scale submarine seal.

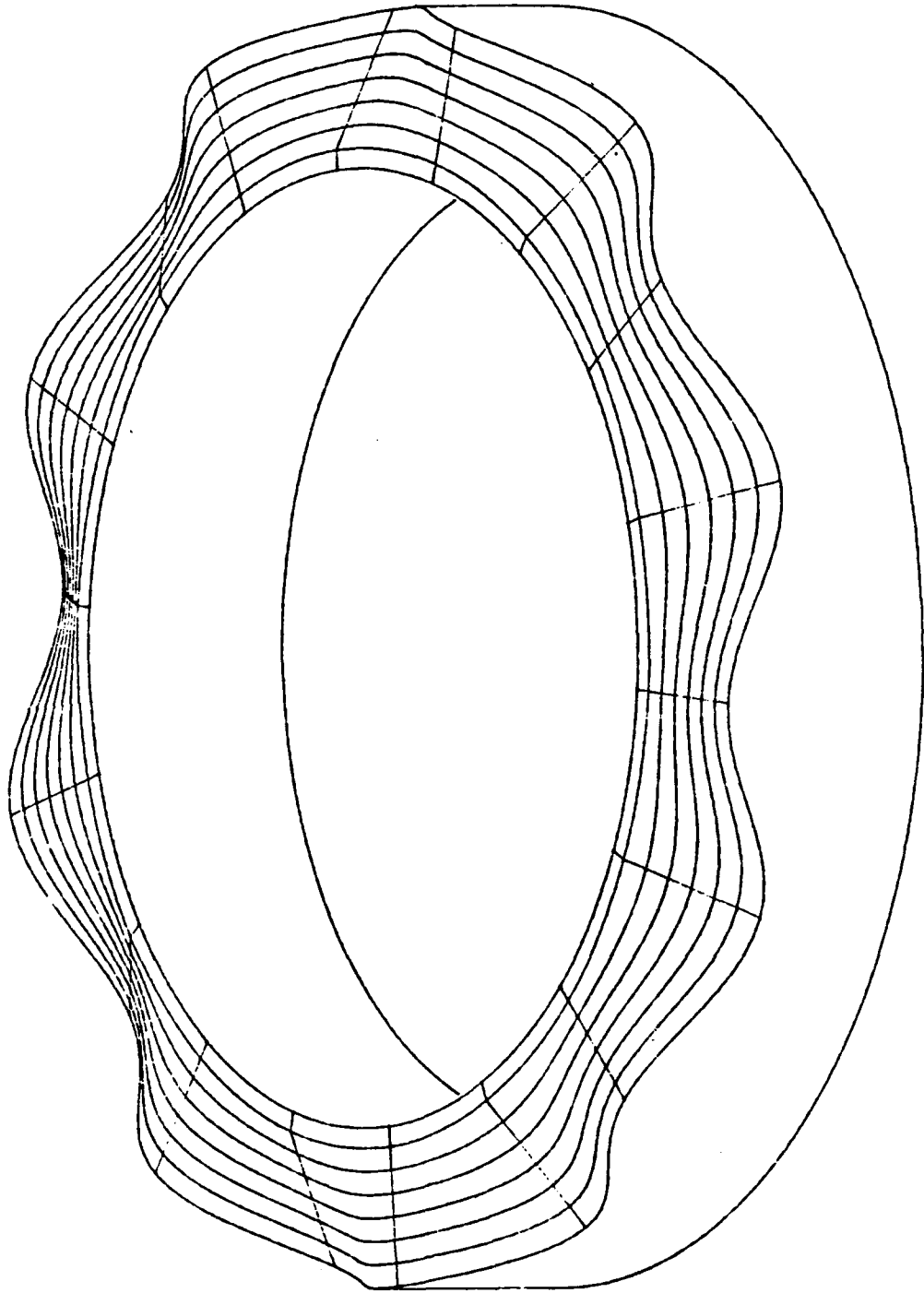


Figure 1-3. Wavy-Tilt Seal Face.
(Patent Pending)

CHAPTER 2

MOVING NINE WAVE SEAL--FIRST DESIGN AND RELATED TESTS

Seal Design Review

The details of the first nine moving wave seal design are given in a previous report [6]. A brief review of the principles of operation will be given here.

Figure 2-1 shows the assembly drawing of the first nine wave seal.

Starting at the left, one of three sinusoidally varying pressures generated by the waviness drive unit [5] is directed into two of six pressure channels (1) in the waviness cylinder (2). The pressurized fluid then passes through the pressure coupler (3) and is ported to one of three circumferential chambers on the left end of the waviness adapter (4). Eighteen smaller passages, connected to each of the three circumferential channels, terminate at pressure pockets (5) located on the inside and outside diameters of the waviness adapter (4). The 54 pressure pockets (three sets of eighteen) apply a waviness pressure to the 54 fingers of the nine wave seal (6).

Figure 2-2 shows a modified isometric cross section of the nine wave seal. The pads shown are labeled (1), (2), or (3) depending on which of the three sinusoidally varying pressures is connected to the pad. The load on the outer pads is directed radially outward and on the inner pads radially inward. With this configuration, three sets of nine waves, each set spaced $13\frac{1}{3}$ degrees from the other, can be elastically imposed on the face of the carbon insert. As discussed

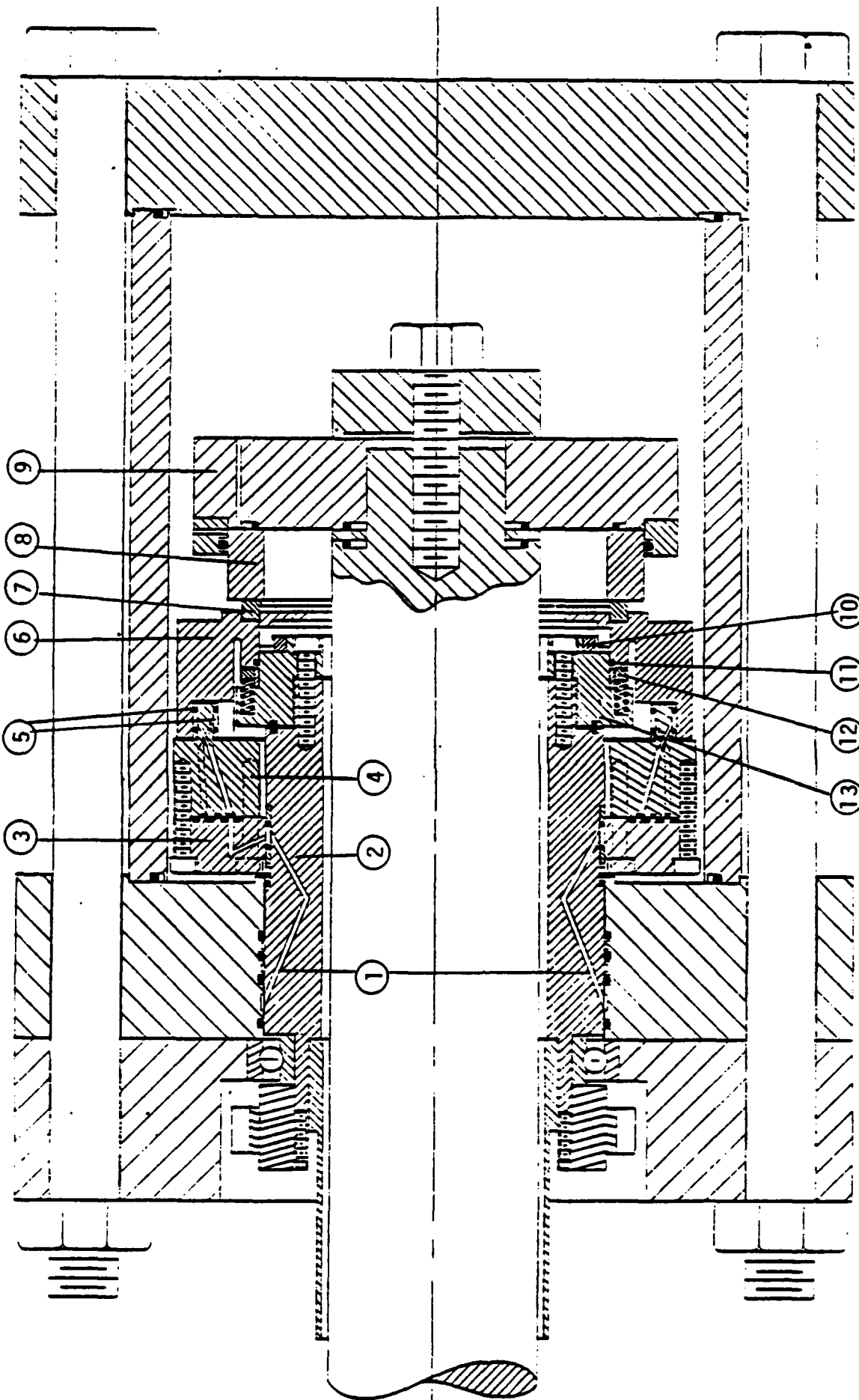


Figure 2-1. Nine-Wave Seal Assembly View.

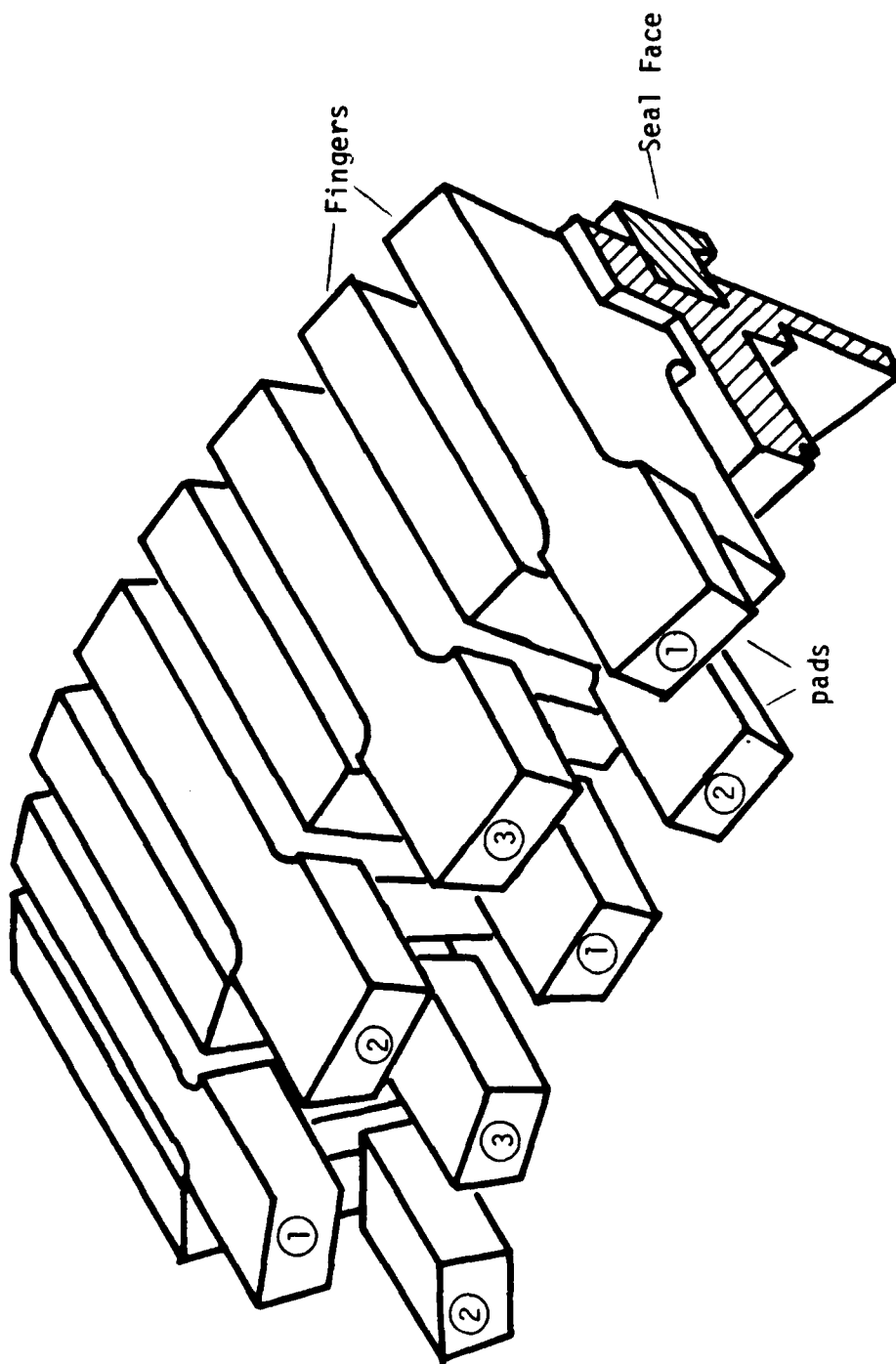


Figure 2-2. Nine-Wave Seal - Modified Isometric Cross Section.

previously [5], by sinusoidally varying the three separate pressures with time, the result is one set of nine waves which moves circumferentially around the seal with time.

The carbon insert (7) is pure carbon P 658 RC material epoxied into the carrier ring using 3M 1838 B/A adhesive. The seat or mating ring (8) is made of silicon carbide. The secondary seal (11) is located at the left end of the seal on the inside diameter. Springs (13) provide preload through the spring seat. Balance ratio is 1.0.

Based on seal modeling reported earlier [5], performance of this seal was predicted as shown in Figures 2-3 and 2-4. A very low wear rate was also predicted.

Initial Test Results

Several initial tests were conducted during which several problems were encountered and resolved. The details are reported in Reference [6]. Based on these early tests, the following changes were made.

- 1) The cross sectional area of the seal was reduced to reduce stiffness and obtain more waviness.
- 2) The epoxy bond thickness was reduced so that the wave would be more effectively transferred from the carrier to the carbon.
- 3) The pressure pad design was modified to use smaller O-rings with miniature pistons.

After these changes were made, performance was much improved, however, there was still some concern over the significant radial taper which had been observed (1500 $\mu\text{m}/\text{m}$ convergent).

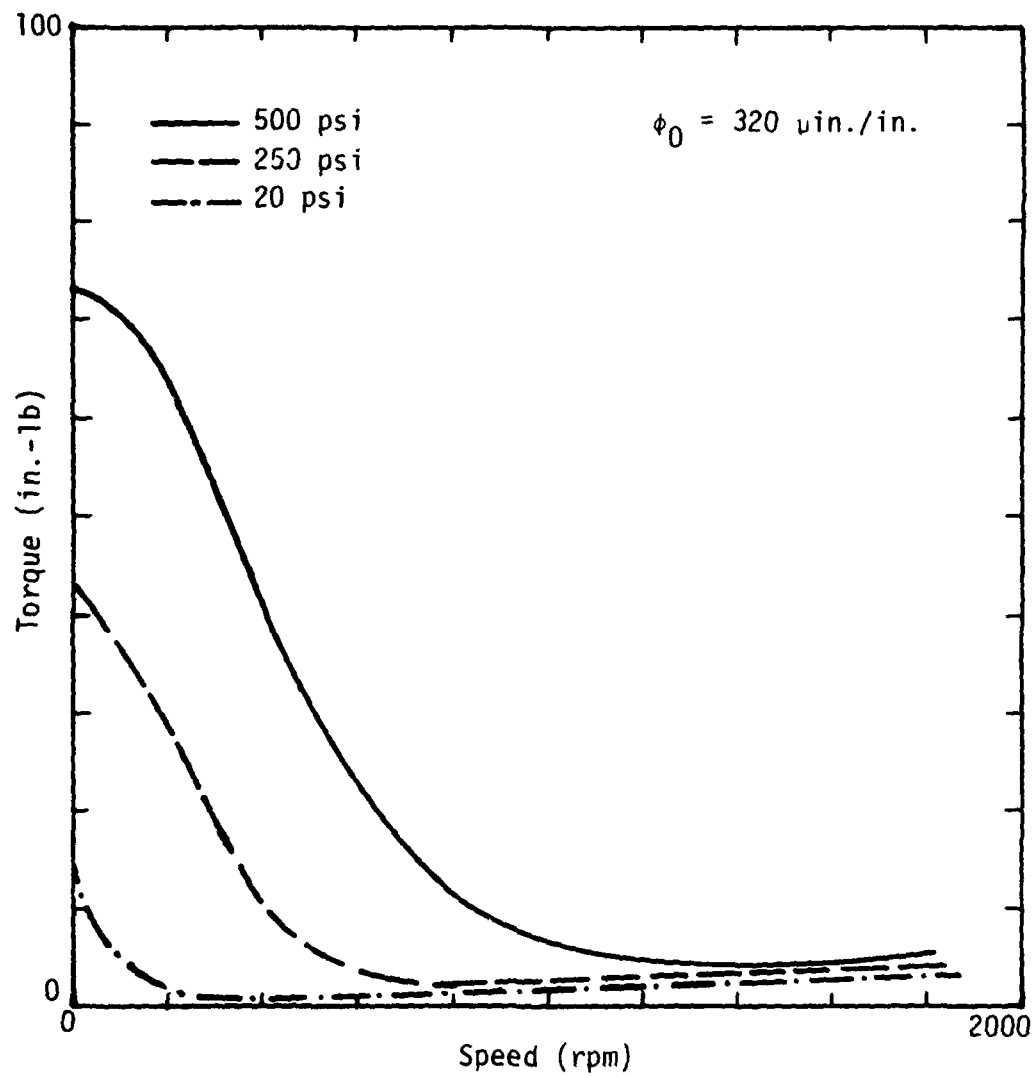


Figure 2-3. Predicted Performance - Torque.

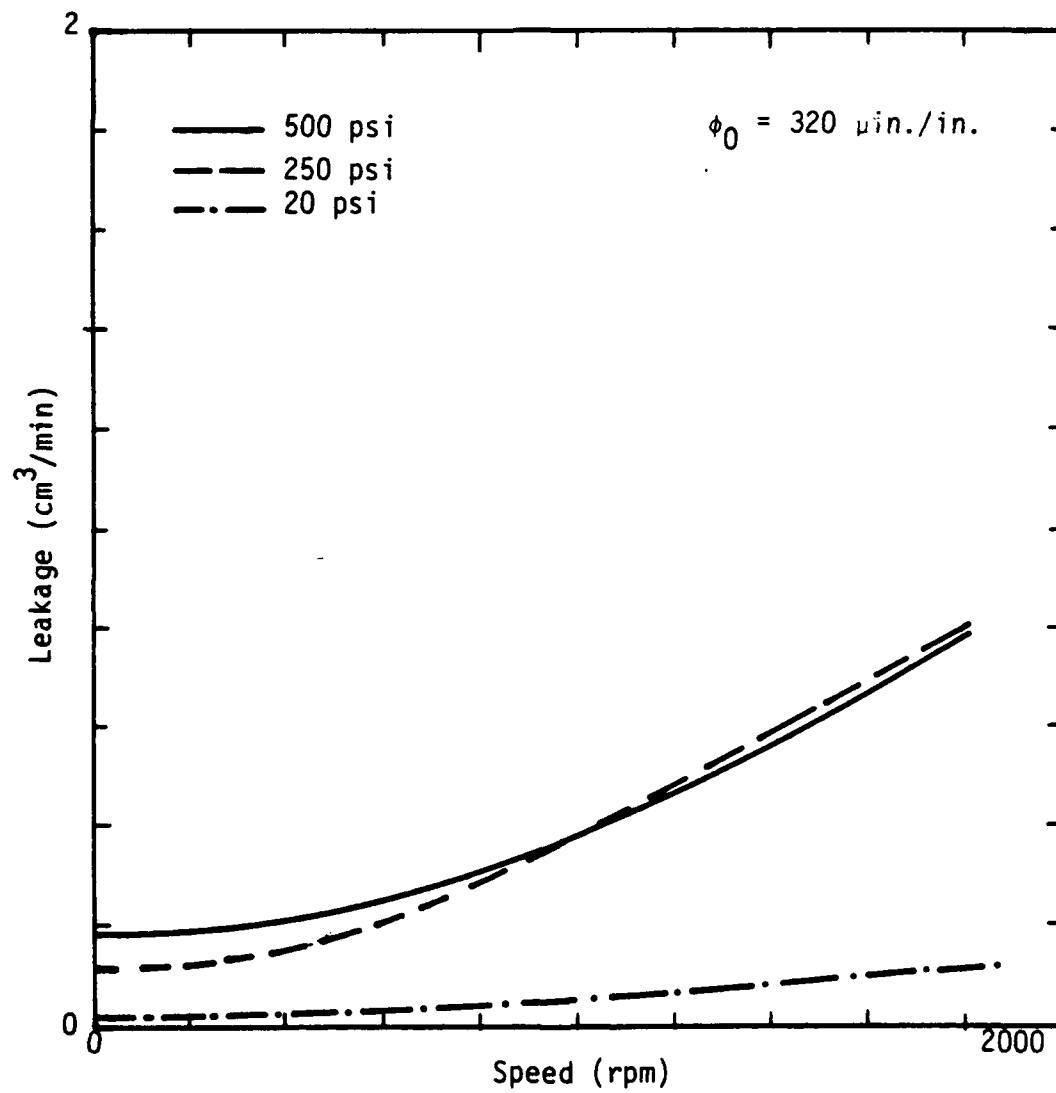


Figure 2-4. Predicted Performance - Leakage.

To determine the possible cause of such a taper several independent tests were run. These were:

- 1) Lap a convergent taper into the seal and test under 100 percent water pressure at low rpm (Test No. 118, Figure 2-5).
- 2) Check for possible creep at end of test.
- 3) Lap the seal flat and assemble the waviness adapter and check for induced radial taper.
- 4) Apply 750 psi to all pads and again check for any induced radial taper.

The results of Test 1 above showed that little if any of the rotation was being caused by pressure induced moments (low speed eliminated thermally caused rotation) as designed. Also, the same radial surface profiles revealed no creep of the carbon insert. Tests 3 and 4 above showed that the pressure adapter and the pressure applied to it does not cause the problem. While the cause is still being investigated, it is pointed out that this taper does not appear to effect the waviness operation. The seal faces are touching all over their surfaces during the test as required.

Tests Nos. 119, 120, and 121 (Figures 2-6 through 2-8) were run to check for software and hardware bugs. These tests were the first to incorporate the full range of variable condition control of the test machine by the computer. Also Test No. 121 was run particularly to check the operation of the combination O-ring-piston pad method of sealing the hydraulic oil. After 168 hours of operation no oil leaks were detected and the test was stopped. With previous hardware and

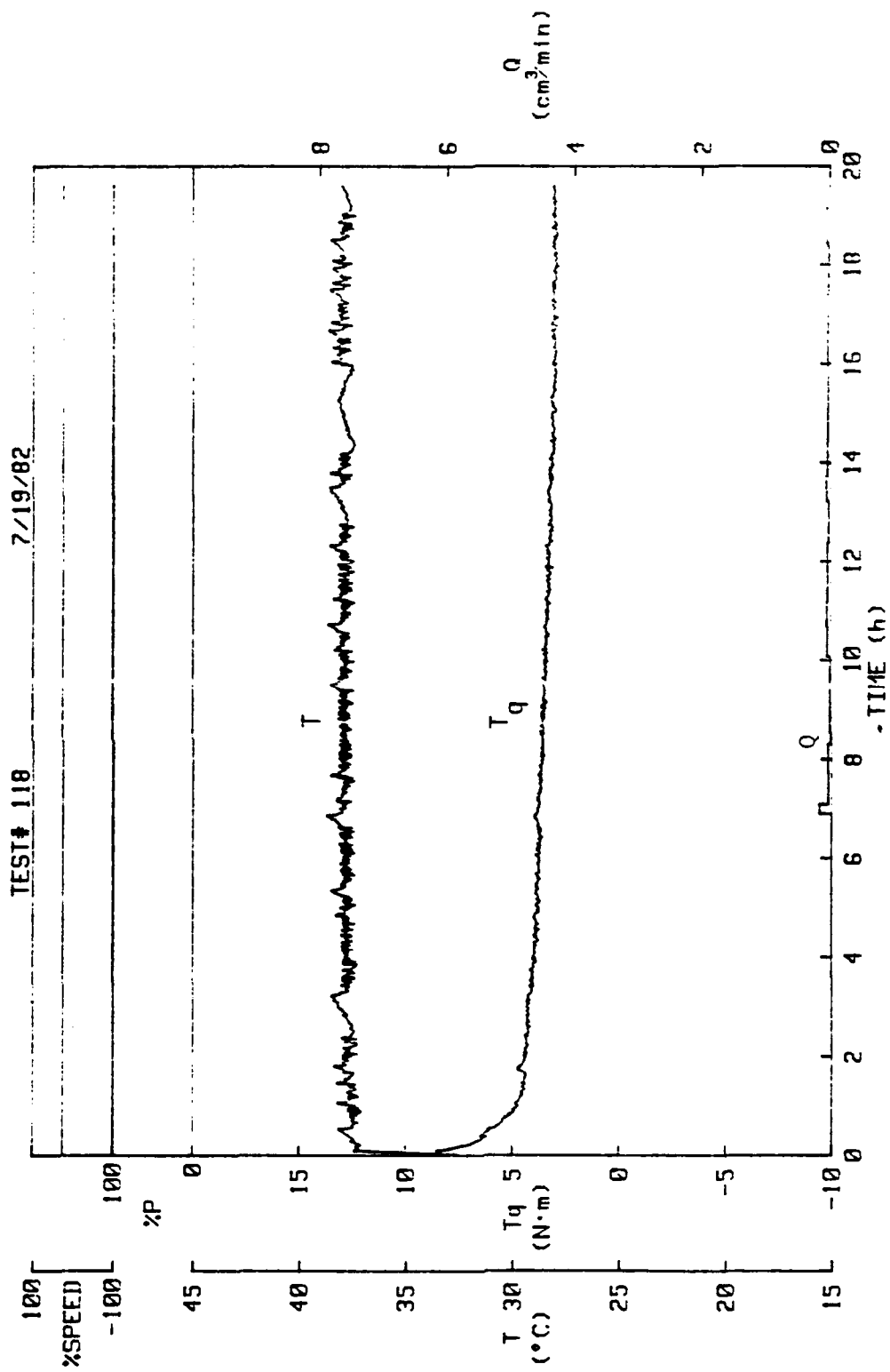


Figure 2-5. Test #118--Nine-wave Seal, 100% Pressure, 40% Speed.

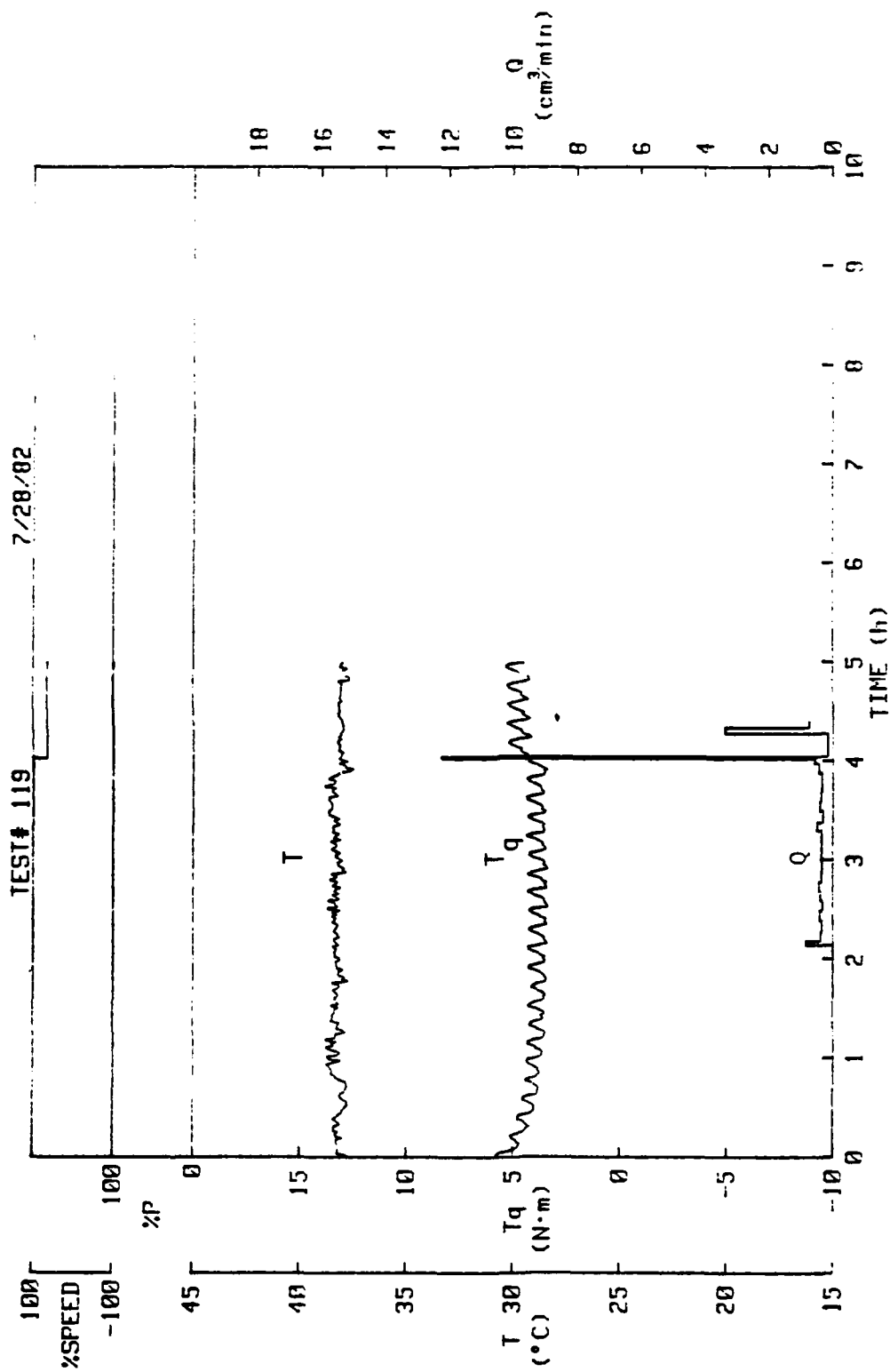


Figure 2-6. Test #119--Nine-wave Seal, 100% Pressure, Variable Speed.

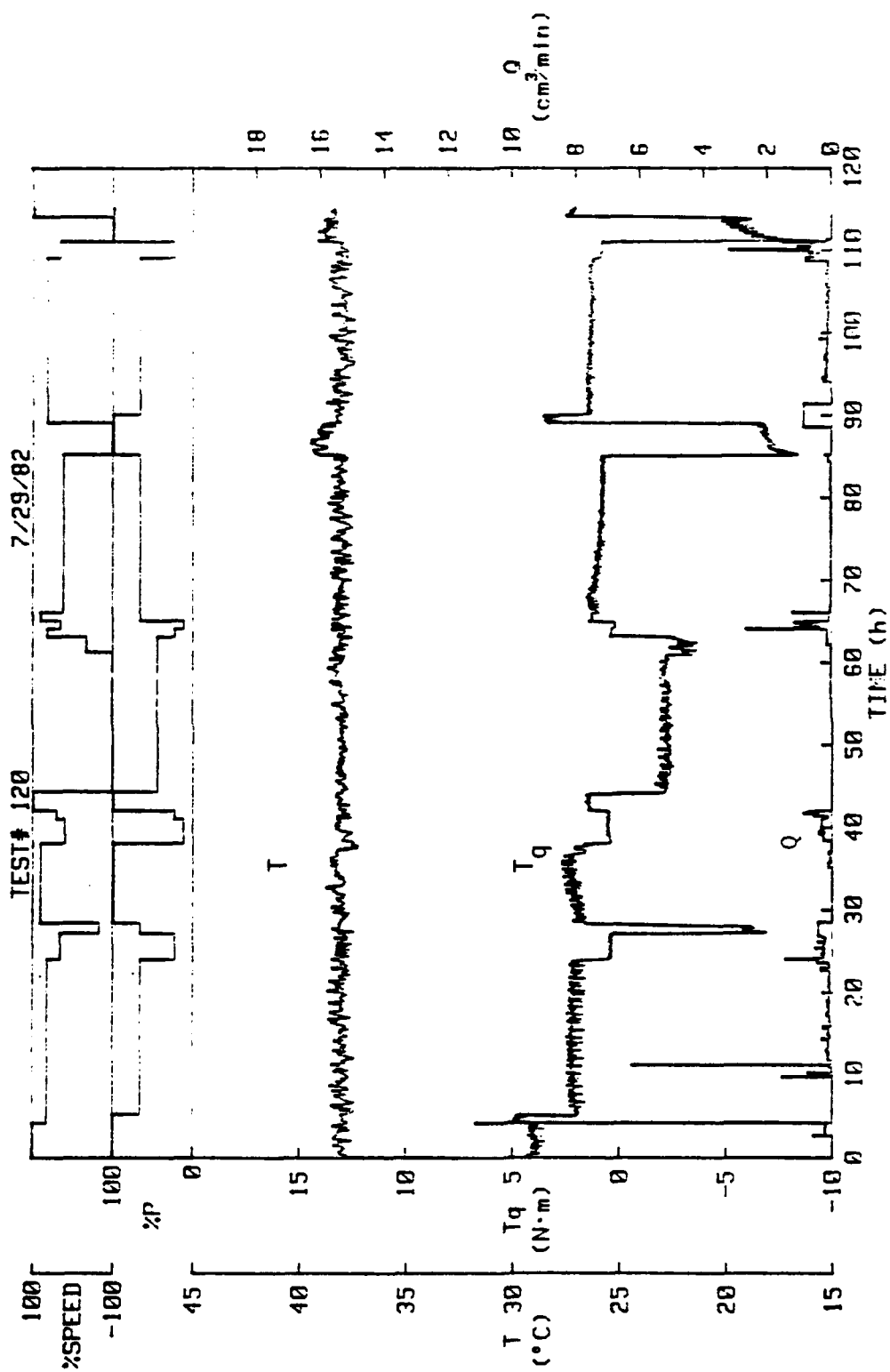


Figure 2-7. Test #120--Nine-wave Seal, Variable Pressure and Speed.

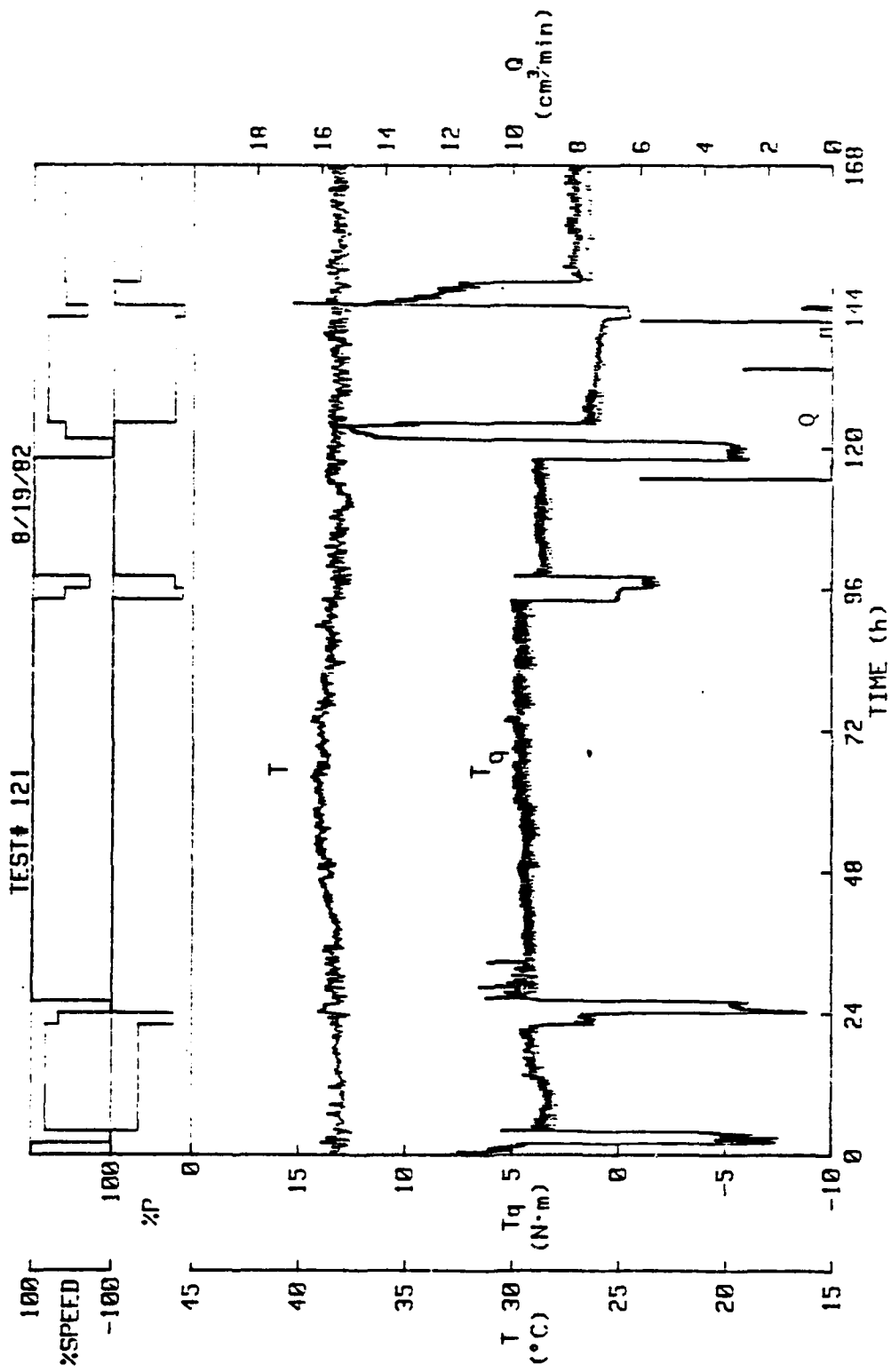


Figure 2-8. Test #121--Nine-wave Seal, Variable Pressure and Speed.

software problems cleared up, the apparatus and seal were ready for the first 2000 hour test.

2000 Hour Test

The seal test period of 2000 hours is similar to Navy seal tests. During the test, pressure, speed, and alignment are varied according to schedule shown in Table 2-1. The cycle period is one week and the summary results shown represent one week of operation. Synthetic seawater maintained at 38°C was the test fluid.

The test apparatus and the variations of operating conditions were all controlled by a computer. Data was recorded and safe operation was assured using the same computer. Details of this system have been reported previously [6].

Performance During Test

Figures 2-9 and 2-10 show seal operation for one week near the start of the test and near the end of the test. The figures show the speed changes and the pressure changes. Seal temperature (taken behind the carbon face), drive torque, and leakage are recorded.

A comparison of the two figures shows that the seal performance becomes much more stable with increased operating time. As an example one of the large changes in torque occurs at 49.5 hours into the weekly schedule (216.5 hours on Figure 2-5). Up to that time torque has been running at near zero under conditions of 7 percent pressure and 14 percent speed for 17 hours. Then within 1.5 hours the pressure changes from 7 to 50 percent and then to 100 percent, the speed remaining at 14

TABLE 2-1
Simulated Submarine Operation Weekly Test Cycle

Hour	Day	% Speed	% Pressure	Tilt & Offset Position
0 —	1	33	100	1
6.5 —		-100		
7.5 —		100		
24 —	2	33	100	3
31 —		14	7	1
48 —		14	50	2
49.5 —	3	-33	100	3
54 —		100		
55.5 —		100		
72 —	4	14	100	3
79.5 —		100	50	2
96 —		Var.	Var.	Var.
102 —	5	100	100	3
120 —		100	100	3
144 —		100	100	3
168 —	7	100	100	3

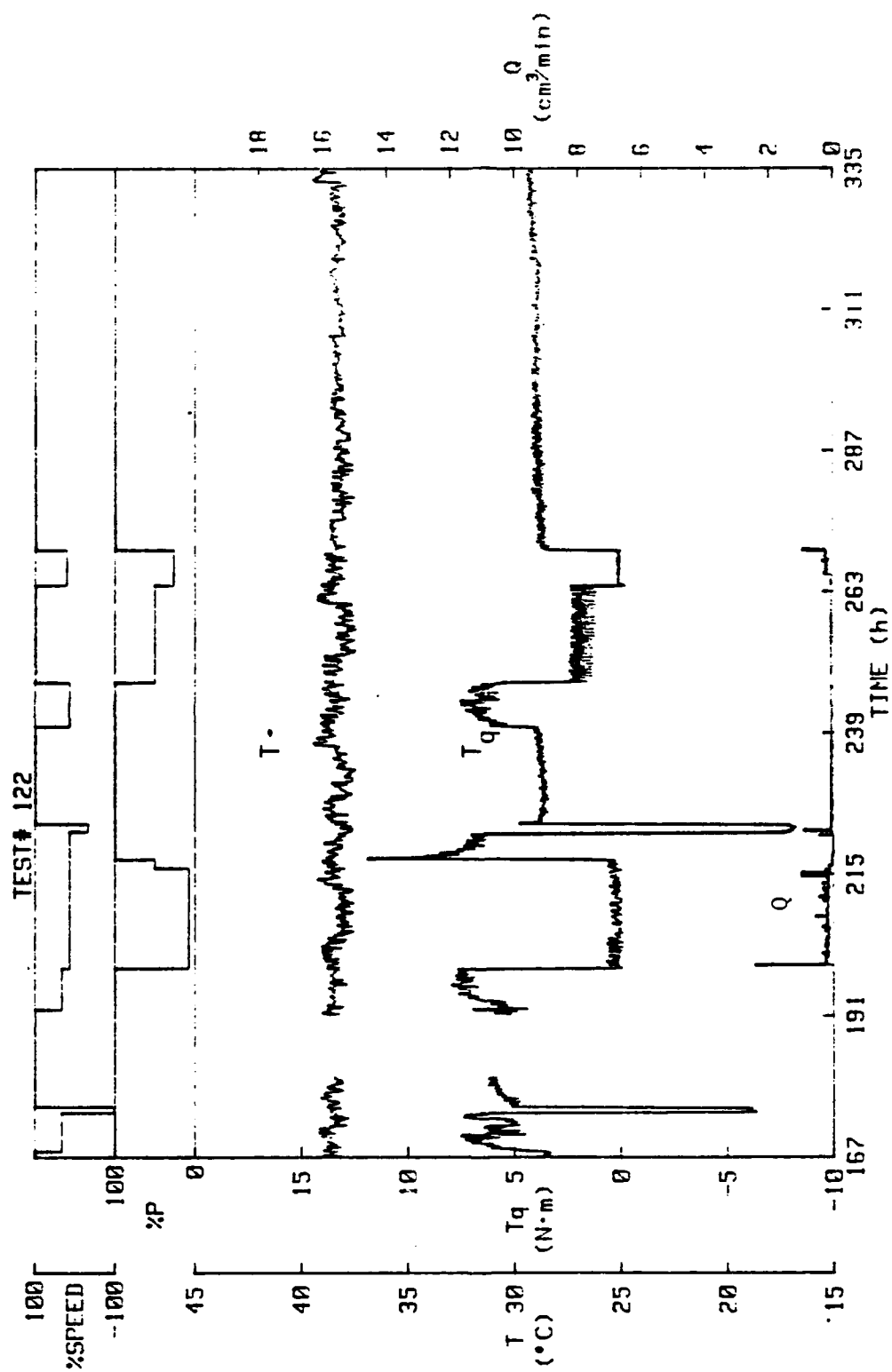


Figure 2-9. Test #122--Nine-wave Seal, 2000-hour Test (initial operation).

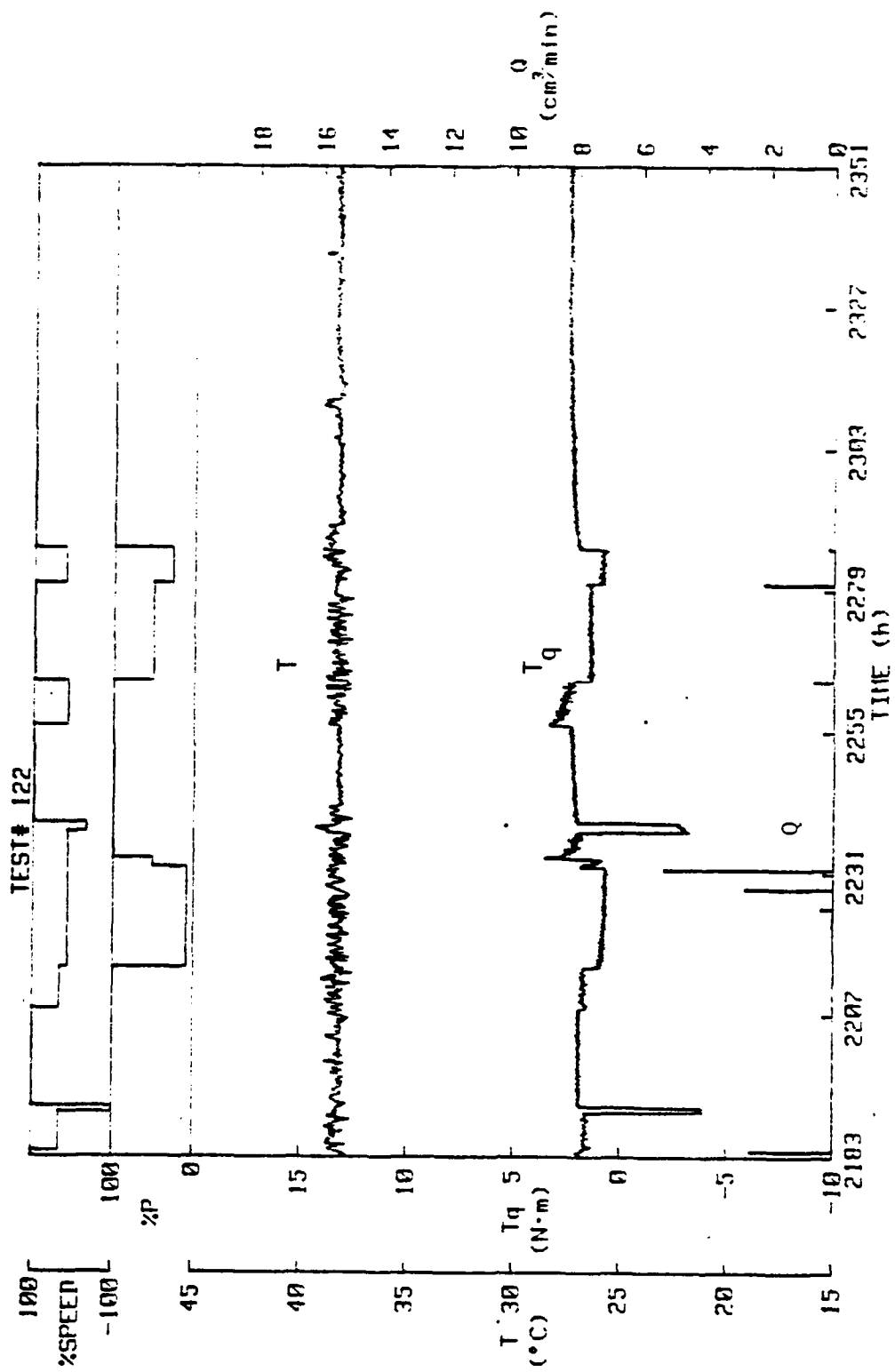


Figure 2-10. Test #122--Nine-wave Seal, 2000-hour Test (final operation)

percent. When the pressure goes from 50 to 100 percent the torque jumps to almost 13 N · m and then starts to fall rapidly. Figure 2-10 shows that this effect is greatly reduced. One possibility for this sudden increase might be that some change in the face profile is occurring causing additional wear momentarily and therefore resulting in a temporary increase in friction. This effect is lessened as the test proceeds possibly because the seal wears to some equilibrium face profile which minimizes these abrupt changes in performance.

Another anomaly in the data is shown by the torque output when there is a direction reverse. Two reversals take place during the weekly program, one at 6.5 hours relative to the weekly schedule (173.5 hours on Figure 2-9 and 2189.5 hours on Figure 2-10) and the other at 54 hours (221 hours on Figure 2-9 and 2237 hours on Figure 2-10). The first direction reversal goes from -100 percent speed to +100 percent speed at 100 percent pressure. On Figure 2-9 the average torque changes from approximately -6 N · m to +4 N · m. The second direction reversal is from -33 percent speed to +33 percent speed at 100 percent pressure. On Figure 2-9, the average torque changes from approximately -8 N · m to +3 N · m.

Figure 2-10 shows that these differences have greatly minimized as indeed they should. The apparent explanation for the anomaly early in the test is that some change in distortion occurs when the direction is reversed which causes some localized hard contact of the faces and higher friction. Evidently this wears away with time and does not appear to degrade performance in the long run.

During the length of the test, the torque transducer was re-zeroed nine times. This was done under zero water pressure by the motor in both the forward and reverse directions and sampling the torque and then averaging the results. These results are shown in Table 2-2. As stated in Reference [6], the torque transducer is sensitive to bearing housing temperature. A temperature increase in the bearing housing causes a torque shift of $0.109 (T_{\text{housing}} - T_{\text{reference}})$ in.-lb/°F. Table 2-2 shows the amount of shift due to temperature effects and the zero shift not attributed to the bearing housing temperature change. The fact that the zero shifted during this long term test might explain some of the differences between torques in the forward and reverse directions at a particular point in time. However, by re-zeroing periodically the uncertainty caused by this error was minimized.

The other measured quantity which changes considerably is that of leakage. At the beginning of the test, Figure 2-9, the leakage was approximately 500 cm^3 for the whole week which gave a weekly average of $0.05 \text{ cm}^3/\text{min}$. By the end of the test, Figure 2-10, the leakage had dropped to approximately 42 cm^3 for the weekly cycle or an average of $0.004 \text{ cm}^3/\text{min}$, a reduction of an order of magnitude. Again, this behavior can be attributed to the seal wearing into a more favorable equilibrium profile.

The leakage data often shows spikes which means that there are spurts of leakage flow being discharged from the test apparatus. There is little doubt that the seal itself does not leak in this fashion; it leaks more steadily. This erratic leakage flow is thought to be caused by some type of trapping, surface tension and wetting mechanism inside

TABLE 2-2

Zero Shift of Torque Transducer

Time (hr)	Zero (N·m)	Temperature (°C)	Zero Shift Attributable to Temperature Change (N·m)	Zero Shift not Attributable to Temperature Change (N·m)
0	33.98	22.04		
29	35.53	25.39	+ .07	+ 1.48
72	34.17	39.67	+ .32	- 1.71
264	33.97	35.71	- .09	- .11
674	34.05	22.28	- .3	+ .38
703	34.72	30.27	+ .18	+ .49
1543	36.81	31.23	+ .02	+ 2.07
1879	37.17	30.85	- .01	+ .37
2047	37.06	28.98	- .04	- .07

the apparatus. Corrosion may also play a role in trapping the leakage flow. Of course, when the speed is changed, a spurt of leakage is expected because the retention of leakage by centrifugal effects changes.

The test was operated a total of 2046 hours. Figure 2-10 shows a larger number because the test was shut down for a period during which the operators were unavailable.

Post Test Analysis

The following observations were made on disassembly:

- 1) The top portions of the pressure coupler and waviness adapter were coated with a light oil film. The waviness drive was started up and run for 3 hours to see if any oil would leak from the O-rings of the waviness adapter. No oil leaks were detected. Thus, there must have been a small leak at some point during the test.
- 2) The seal face temperature thermocouple was slightly worn due to rubbing contact with the pressure vessel.
- 3) Rust deposits were found on the threaded ends of the bolts used on the spring seat.
- 4) Considerable amounts of mineral deposits (white scale) were found on the inner portion of the seal where $p = 0$ psi.
- 5) The seal face was polished with some mineral deposits on the inside diameter edge.
- 6) The Delrin spacer used with the spring retainer experienced enough swelling to make removal difficult.

- 7) The waviness cylinder could not be rotated fully to Position No. 1 or No. 3. This had become a problem during the last 500 hours of operation such that the full extent of misalignment could not be achieved (see misalignment mechanism described in [6]).
- 8) The small O-rings on the pressure adapter received about a 0.002 in. permanent set.
- 9) Some chipping was noticed on the outer edge of the $K_T\text{Si-C}$ seat.
- 10) Steel end plates showed considerable rust and corrosion on bottom half of the inside surface.
- 11) The worm gear had severe rust deposits on the lower 25 percent of it.
- 12) The inside diameter of the torque transducer at the front end (end plate side) showed signs of galvanic corrosion.
- 13) The bearing which fits over the waviness cylinder was removed from the end plate and disassembled. The bearing was locked due to rust and corrosion between the balls and races.
- 14) The waviness cylinder and other seal parts showed no corrosion.

The carbon ring was measured for wear and radial taper. Figure 2-11 shows a typical radial surface profile showing the unexplained radial taper (which averages 2700 $\mu\text{m/m}$). The profile also shows that contact occurred across the entire face.

Table 2-3 shows a summary of the wear values taken at five different locations around the seal. The technique for making wear

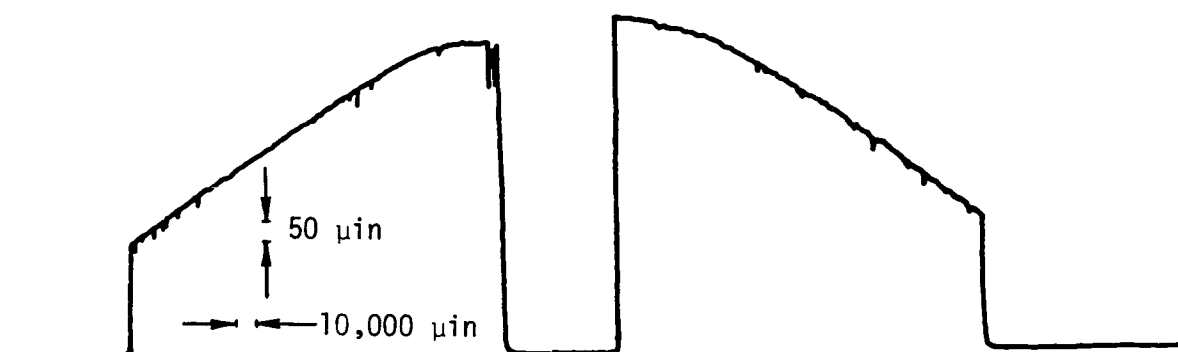


Figure 2-11. Final Radial Surface Profile of Carbon, 2000-hour Test.

TABLE 2-3

Wear Measurements

2000 Hour Wavy Test

Comparison to 8/18/81 Traces--Test #122

Location ¹	Original Nose Height (μ in)	Final Nose Height (μ in)	Wear (μ in)
1-6 ²	3925	3700	225
4-5 ²	3900	3050	850
2	3950	2750	1200 ³
4	3800	3100	700
6	4000	3650	350
Average excluding Location 2			530 μ in 13.5 μ m

¹ Because of deposits on wear track, not all locations could be measured.

² Designates a trace taken midway between the given locations.

³ Questionable value because of deposits.

measurements is to initially grind a very shallow and narrow offset (0.004 in.) at the seal ID. The original height of the face relative to this offset is determined using a profilometer as shown in Table 2-3. Then, after the test is completed, the same measurements are made and compared.

One problem using this method is that sometimes unusual deposits form in the offset, invalidating the wear references. This problem was taken into account in compiling the data in Table 2-3. The wear on the Si-C ring was just a few microinches.

Concerning the general character of the test results compared to prediction (see previous figures) torque values are somewhat higher and leakage values are lower. It was concluded that given the problem in applying the waves as discussed that the wave amplitude is not as large as the values used in the calculation and that further comparison would not be too meaningful. However, a detailed comparison is made for design 2 in Chapter 3.

Conclusions on First Nine Wave Design

1. The moving wave concept works in actual operation. That is a moving wave was imposed which caused low friction and wear while maintaining low leakage. Earlier tests and later tests using the same seal in a parallel face mode show that drive torque is several times higher in the absence of waviness.

2. This first nine wave seal design is very difficult to make. Its design requires many intricate cuts and small holes to be made in relatively hard to machine Inconel 625.

3. The hydraulic drive for the pressures for the moving wave, while being a relatively simple device, does add a layer of complexity to the seal design. A digitally controlled device might in fact be simpler than the mechanical device used.

4. The use of the oil hydraulic system to drive the wave can be accomplished reliably on a 2000 hour test. However, as discussed in the previous report [6], the utmost care in O-ring seal design must be used to eliminate O-ring wear and leakage. There is some question as to the long term (10 years) reliability of a device such as this although hydraulic system of other types in known applications do operate reliably for many years.

5. While wear and friction were quite low, they were expected to be lower. The reason for this is that the actual design value of wavy tilt could not in fact be imposed on the seal. Just over one half of the design values was available because of the limitation of the mechanism as discussed previously [6].

6. The corrosive problems mentioned previously were all related to a design change which allowed leakage to flow on occasion into the gear mechanism of the waviness cylinder drive--a region not designed for water contamination. This corrosion also caused the binding in turning the waviness drive being used as the misalignment mechanism. This problem was readily overcome for future tests by redesigning the leakage flow path.

7. The seal and support system were made using inconel, 316 stainless and monel. Concerning corrosion on the seal and supporting parts, there was no detectable (to the eye) corrosion of surfaces nor

galvanic corrosion between mating parts (except the screws noted). The electrical insulation method of mounting the parts made of different metals apparently was adequate.

8. The bond between the carbon and its support or holder is very critical. Early experiments show that a thick (0.010 in.) epoxy bond will permit the wave which is imposed on the carrier to simply be flattened out after it reaches the carbon. A very thin bond, which has a much greater stiffness, was used to solve this problem.

9. After an initial wear in, performance becomes consistent from week to week. There was no degradation of performance over the course of the test. The seal had lower friction and lower leakage during its last week of operation than during the first week.

10. The total wear was low and could be expected to meet ten year life objectives. Projecting the average wear of 530 μ in. to 150000 hours of continuous operation gives a total wear of 0.040 in., well within what can be allowed for in a seal design.

11. The radial taper source has still not been identified. Some type of distortion which has not been duplicated in bench tests is occurring in the actual seal installation itself. Some of the wear is being caused by this behavior. In the test, however, once beyond this wear in period, the radial taper does not appear to be affecting the results.

500 Hour Flat Face Test

Test No. 123 was run using the nine-wave seal but without any waviness, a flat face test. The conditions of operation were the same

as for the 2000 hour test so that the performance could be compared for the two. Figure 2-12 shows a sample weekly plot for the flat face test. Comparing the performance at 100 percent conditions reveals that the torque in the flat face test was nearly four times higher than in the waving test (approximately 9 N · m to approximately 2 N · m). Thirteen times during the test the test machine was shut down by the control system because of excessive horsepower demands due to high friction. Torques as high as 46 N · m were recorded. Test No. 123 was actually operated for a total of 420 hours and then stopped.

Wear measurements were taken for the flat face test so that a comparison could be made with the wavy test. Final profiles showed an average wear of 117 μ in. (3 μ m) for the 420 hours of operation (Table 2-4). Adjusting this to the 2046 hours for the wavy test would give a wear of 570 μ in. (14.5 μ m). This is essentially the same; however, the comparison is not really valid.

The 100 percent pressure and speed conditions did not represent the proper fraction of the total time of operation because of the high amount of down time at 100 percent conditions. The actual time operated at 100 percent conditions was 186 hours of the 420 whereas it should have been 317 of the 420. From this information, the wear rate could have been as much as two times higher had the test run properly.

In conclusion, the wavy seal has a much lower drive torque and probably a lower wear rate. Leakage of the parallel face seal is lower although the wavy seal leakage is also quite low. Probably the most important but hard to evaluate difference is that the flat face seal has torque spikes (none are shown in Figure 2-12, but they do occur).

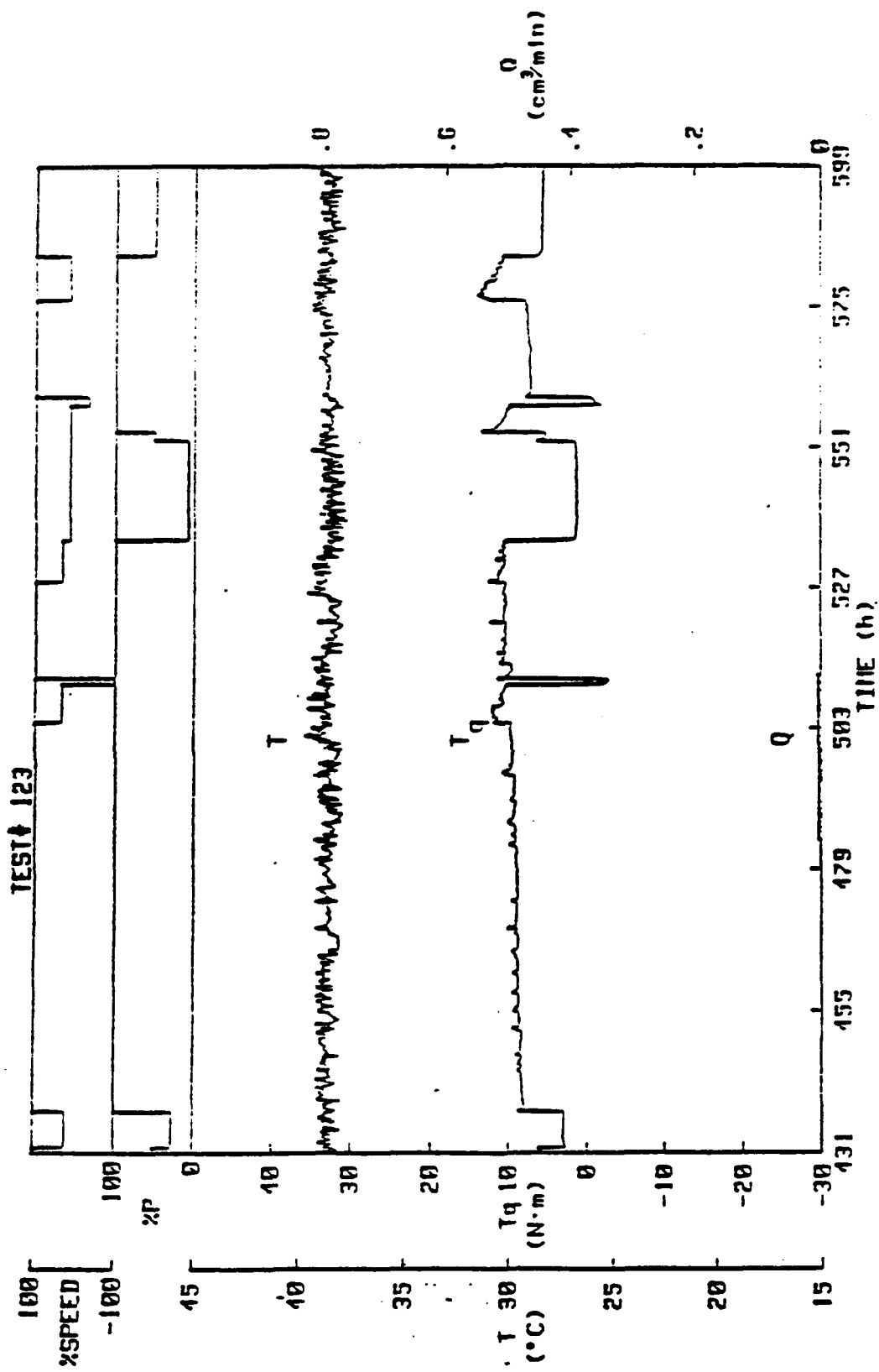


Figure 2-12. Test #123--500 Hour, Flat Face Test, Variable Conditions.

TABLE 2-4

Wear Measurements--500 Hour Parallel Face Test

Location	Wear (μ in)
#1	200
#4	150
#3	200
#6	50
#2	0
#5	100
Average	<u>117 μin</u>

The wavy seal does not. It is thought that the torque spike is caused by divergence and pinching off fluid pressure load support across the film--thus high friction. High average torque and torque spikes represent potential problems in that they cause high mechanical loads on the system, larger thermal loads, and the possibility of causing thermal instability. Consistent nonsticking drive torque is preferable.

The wear results above may be interpreted as suggesting that at this level of PV service the carbon-Si-C wearing pair may give adequate seal life. In many industrial applications where face wear is the limiting factor this is the case. However, for higher PV and for cases where extremely long life is of interest, then reduction of face friction and mechanical contact, such as by using a wavy seal, is very important.

Integral Force Transducer Design and Test

After evaluating the previous design, many alternatives for a simpler design were considered. As part of this evaluation, it became clear that a different type of force transducer should be considered which might make implementation of other types of wavy seal design much simpler. The criterion for the force transducer were:

- 1) Must convert pressure to force proportionally and consistently.
- 2) Must be absolutely reliable, i.e., no leaks or blowouts permitted.

- 3) Must be salt water compatible.
- 4) Must be easy and low cost to fabricate.
- 5) Force must be relatively insensitive to clearance (otherwise a serious tolerance problem and other uncontrollables result).
- 6) Must allow force to consistently go back to zero.
- 7) Must allow for easy manifolding of pressure.

To meet these needs, the idea shown in Figure 2-13 was developed. The cylindrical steel shell is for the purpose of testing the concept. The epoxy could be formed around many transducers and their tubes in any holder as needed. The idea is that pressure will cause fluid to flow from the tube into the porous metal. The porous metal is contained on all sides but the diaphragm side by epoxy. Since the diaphragm is thin, the fluid will raise it until it touches a mating part as shown in Figure 2-14. Then the fluid will become pressurized behind the diaphragm and apply the needed force.

The epoxy casing and diaphragm make the device corrosion proof. The tubing connections will all be encased in epoxy so that manifolding can be performed using simple soldered connections. Cost would be low once initial fixtures were fabricated.

Several such devices were fabricated as shown in Figure 2-13 and tested as shown in Figure 2-14. The results are shown in Table 2-5 and Figure 2-15. Figure 2-15 shows the measured force for one cycle of pressure. Force is clearly linearly proportional to pressure. Test 1 was the first successful transducer. Tests 2, 3, and 4 show how this transducer operated with increasing clearance. As expected the force decreases with increasing clearance. For the 0.5 in. diam. unit the

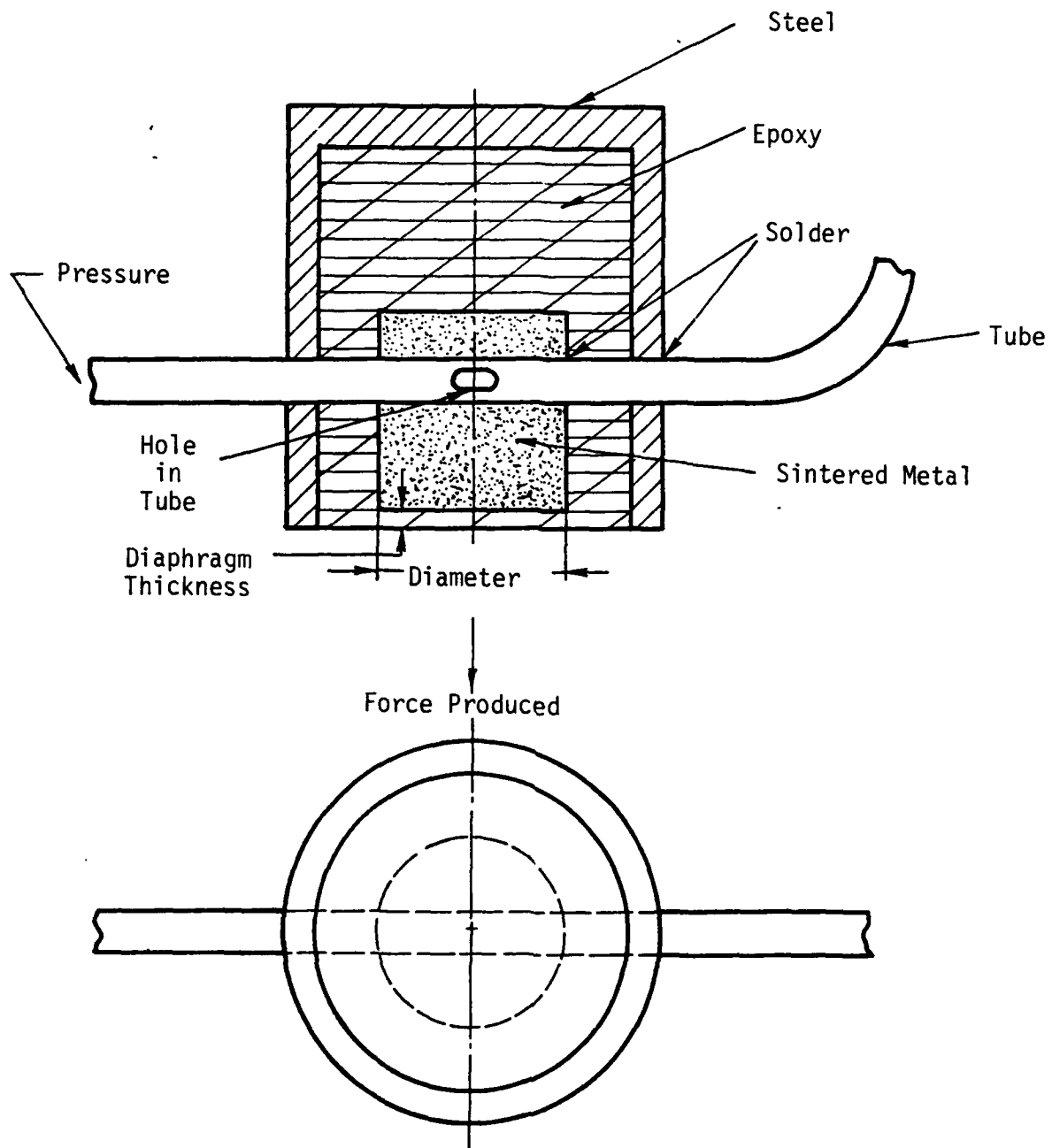


Figure 2-13. Integral Force Transducer.

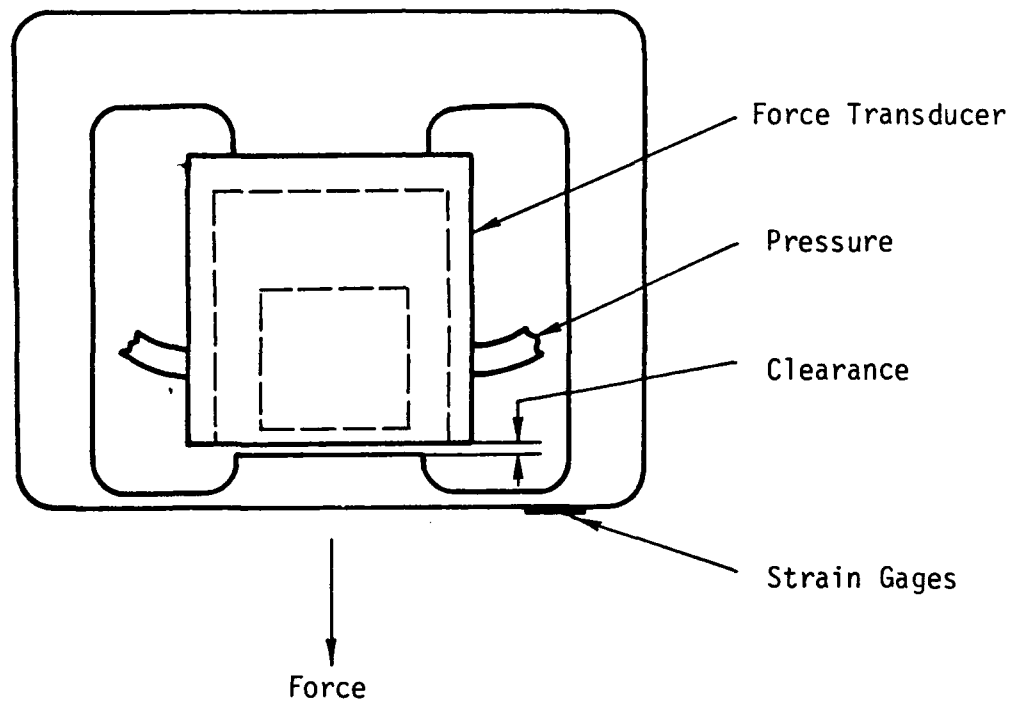


Figure 2-14. Integral Force Transducer Test Setup.

TABLE 2-5
Tests of Epoxy Force Transducers

Test	Element	Diaphragm	Clearance
1	0.5 in dia. sintered	0.021 in	.0 in
2	0.5 in dia. sintered	0.021	.0013
3	0.5 in dia sintered	0.021	.0024
4	0.5 in dia. sintered	0.021	.0046
5*	0.5 in dia. sintered	0.021	.0046
6	0.25 in dia. sintered		.0
7	0.25 in dia. sintered		.0015
8	0.25 in dia. sintered		.0030

*Run for two hours

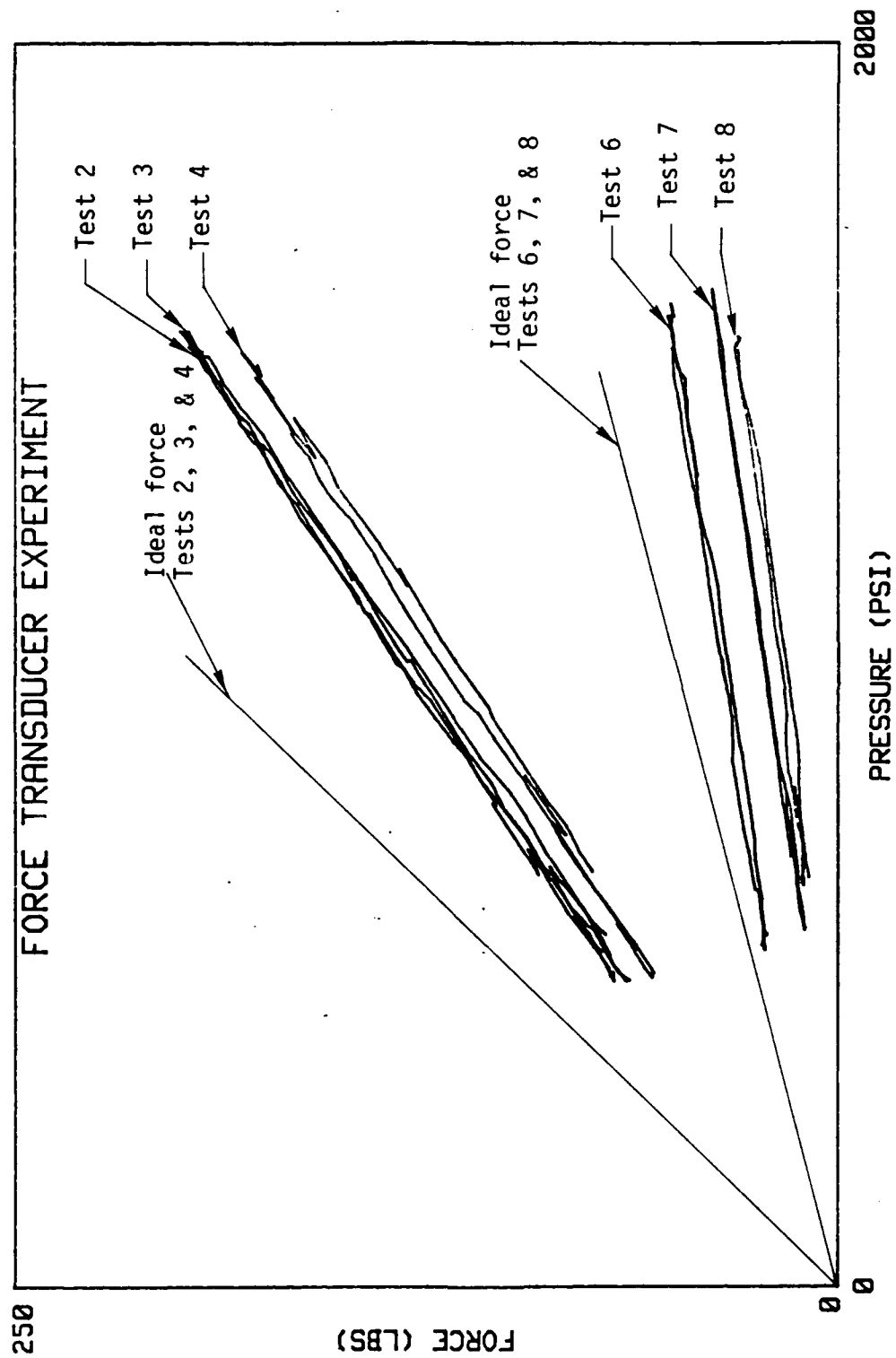


Figure 2-15. Integral Force Transducer--Force Test Results.

percent reduction would be considered acceptable for a seal design. From Tests 6, 7, and 8, for the 0.25 in. diam. transducers, the percent reduction is much larger, indicating that a thinner diaphragm should be used to decrease the sensitivity. Test 5 (not shown) was cycled for a period of two hours to evaluate endurance of the device. The comparison of the actual force produced to the ideal force shows that the epoxy diaphragm is holding back a substantial fraction of the pressure load.

While force performance above was considered to be satisfactory for the initial attempts to create such a device, the reliability of the device came into question. Two of the devices ruptured on first use and one ruptured after being used in multiple tests. While some of these problems were related to fabrication technique, it became clear that epoxy may not be the most reliable material for a pressure containment application. A material with more reliable strength and flow characteristics was needed.

At this point the study was ended as it was decided to use traditional O-rings with pistons for the next design. The logic was that O-rings with proper piston were known to be reliable, and while not being necessarily the best force transducer from a fabrication standpoint, could be applied with greater expediency than finding and developing the proper candidate material for the integral force transducer. Even so, the integral force transducer work is encouraging should some application warrant further development.

CHAPTER 3

NINE WAVE SEAL--SECOND DESIGN

Criteria

Even though the performance of the first nine wave seal design was reasonably good, it was decided that performance should be somewhat better if full waviness could be applied and, to make the wavy seal practical, a simpler, more reliable means of applying the wave must be designed which can be more easily fabricated. Thus, the overriding criteria were simplicity, fabricability, and reliability.

The specific criteria for the design of a wavy seal are as given previously [6]. They are:

- 1) Very low wear--10 year life.
- 2) Moderate to low leakage--leakage consistent over time.
- 3) Low friction to ensure low thermal distortion.
- 4) Operation in the 500 psi, 1800 rpm at 4 in. mean diameter range.
- 5) Operation in seawater.
- 6) Seal components themselves must be reliable to be compatible with 1) above.

To design a wavy seal, additional criteria must be satisfied for proper operation.

- 1) The seal rings must be of a zero-moment design, i.e., no rotation of the rings due to changes in the sealed pressure should occur.

This eliminates the need for the seal rings to wear to a new profile at each operating pressure.

2) Nine waves must be imposed on the seal. Reference [6] shows the need for nine waves so as to increase the relative tilt stiffness of the seal and thereby maintain a face profile which is relatively unaffected by operating conditions. Thus, different operating conditions do not wear new profiles on the seal.

3) Application of the waviness forces must be made to the zero pressure side of the seal. This eliminates the need to compensate for the effect of sealed pressure on the applied waviness force.

4) The centroid of the cross section must be optimally located to provide a continuous sealing dam around the seal to give both minimum leakage and maximum load support.

5) Stiffness of the seal ring is to be minimized. This will ensure compliance of the seal ring with the mating ring at lower harmonics and also makes application of waviness easier.

Initial Configurature

Figure 3-1 shows the initial basic configuration used in the design analysis. There are 54 pistons located circumferentially around the inside diameter of the metal ring. Three sets (18 pistons per set) are pressurized with a sinusoidally varying hydraulic pressure. The 18 pistons per set are divided up so that there are nine pistons acting to the left of the centroid and nine pistons acting to the right. This produces one set of nine waves. The three sets of 18 pistons are connected as in the previous design to provide nine moving waves. The

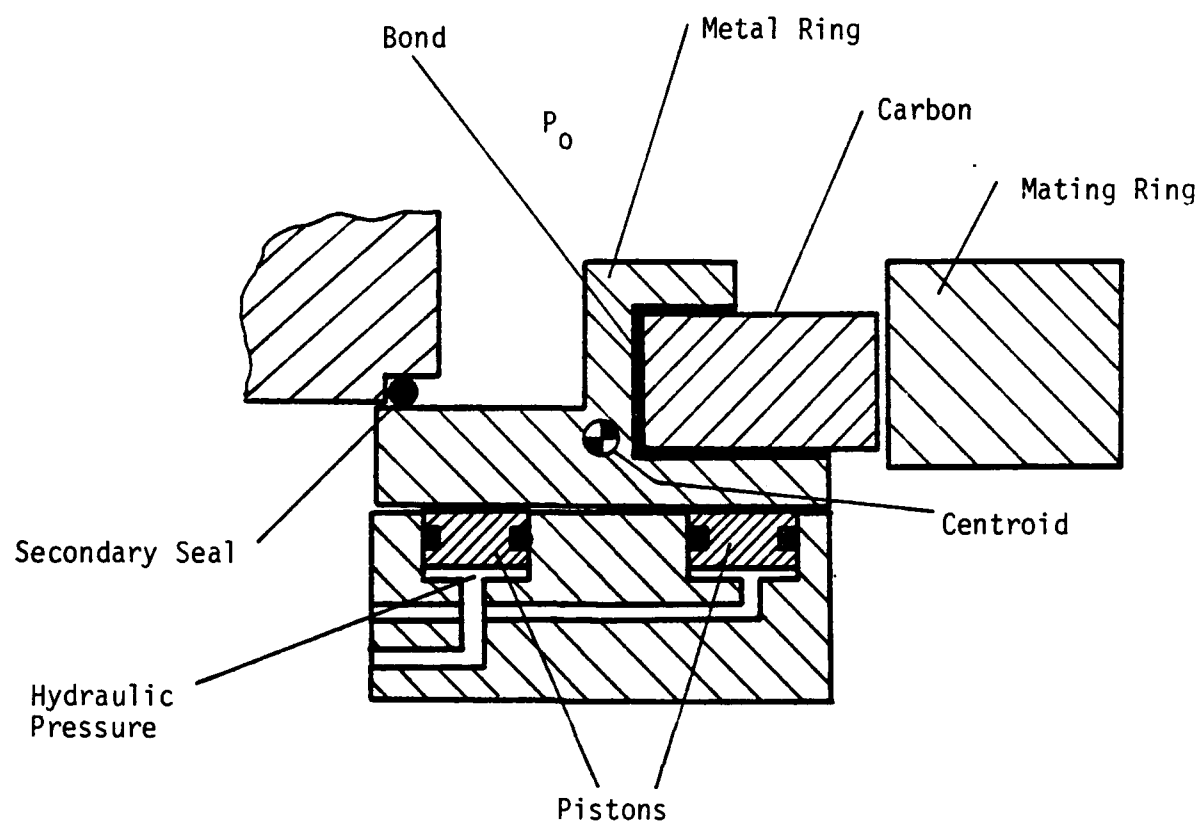


Figure 3-1. Initial Nine-Wave Seal Configuration - Design 2.

carbon is bonded as before into the metal ring which is made of Inconel 625. Inconel 625 was chosen for its excellent corrosion resistance in salt water.

The configuration shown avoids some of the problems associated with the previous design. The pistons are considered to be a more reliable method of applying the forces with no leakage. Manifolding of the pressure can be accomplished with small diameter tubing (see detailed descriptions later). The entire force mechanism is located on the leakage or zero pressure side of the seal so that the mechanism is not affected by operating pressure. On the other hand this arrangement does cause a high hoop stress as discussed later.

Seal Ring Design Solution

Figure 3-2 shows the details of the seal ring. The fixed parameters are essentially the same as for the previous design [6]. The method of solution is identical except for the location of the pressure force that produces the waviness and the method of zeroing out rotation due to sealed pressure. In this new design obtaining a zero moment can be easily achieved by adjusting the secondary seal position.

In addition to meeting the previous criteria, it was decided for this new design that the product GJ_0 (torsional stiffness) must be about ten times lower than for the previous design. This condition was imposed to ensure that the needed wave could be produced at a reasonable hydraulic pressure and moment arm. Due to this requirement, the metal part became quite thin. This means that even though $E_{\text{carbon}} \ll E_{\text{metal}}$, the carbon in conjunction with the metal tends to stiffen it up

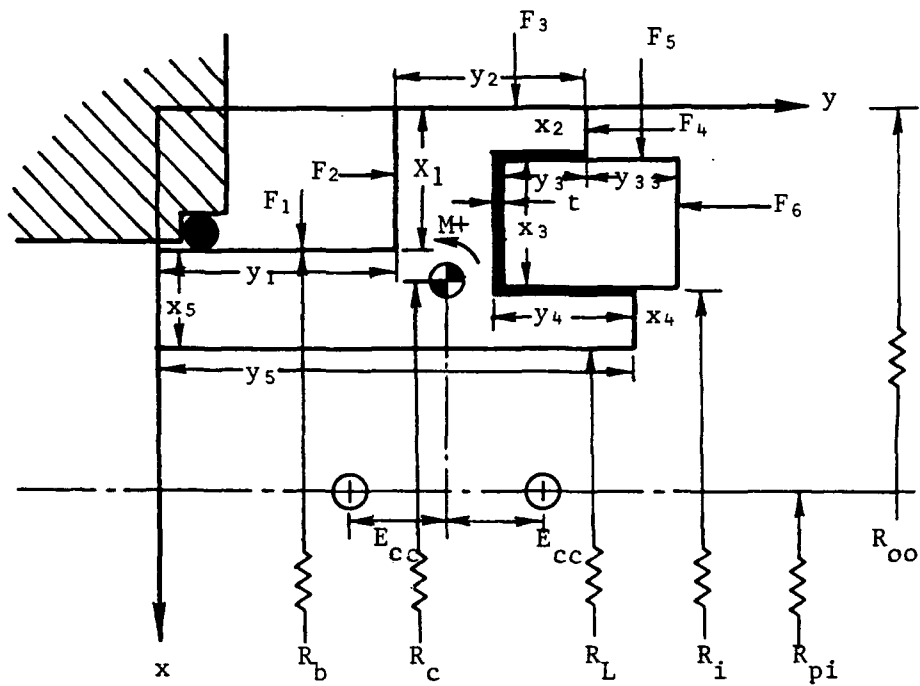


Figure 3-2. Composite Seal Ring Geometry.

and one cannot use the metal only as a basis for computing section properties.

Bending Stiffness

To get an estimate of the stiffening effect of the carbon, the cross section shown in Figure 3-3 was analyzed using a FEM program. The cross section shown is typical of those analyzed as having potential for the final design. The ring was analyzed in bending as produced by a moment about the circumferential axis uniformly distributed around the ring. The metal and the carbon were considered to be perfectly bonded at the interface.

Based on the rotation of the cross section in the plane of the figure as predicted by the FEM program, the equivalent stiffness of the composite ring was computed using the formula for twist (bending) of a ring due to a uniform moment about its circumferential axis.

$$\phi = \frac{m_{\theta} r_c^2}{EI_x} . \quad (3-1)$$

The FEM prediction is:

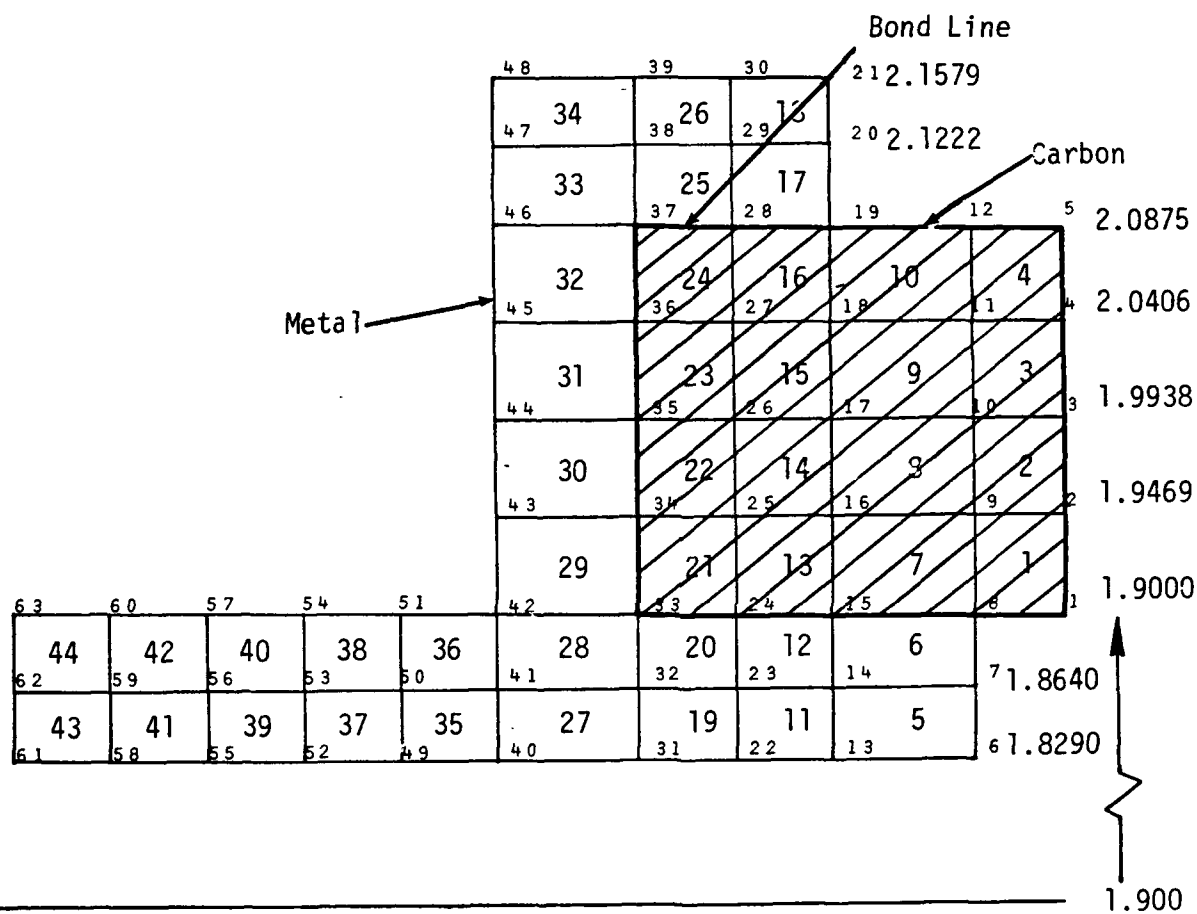
$$(EI_x)_{\text{equivalent}} = 5.8 \cdot 10^4 \text{ lb in.}^2 . \quad (3-2)$$

Based on the metal only

$$EI_x = 2.3 \cdot 10^4 \text{ lb in.}^2 . \quad (3-3)$$

The conclusion is that the composite section is about twice as stiff in bending than as predicted by considering the metal only. This

.5355
.4854
.4353
.3852
.3351
.2850
.2317
.1783
.1250
.0600
0.0000



added stiffness is particularly important when calculating out of plane buckling or conformability of the ring and was taken into account in the design.

Buckling

For rings of small cross section under external pressure, buckling of the ring must be considered. Such a ring can buckle in the plane of the seal face or it can possibly buckle out of the plane. The second case, while unlikely, appears to be the more limiting case for the type of ring of interest so has been used as a design criterion.

Williams [30] treats the non-symmetrical cross section ring buckling problem and shows that for out of plane buckling the critical radial load can be predicted using the following equation.

$$C_1 \bar{p}^3 + C_2 \bar{p}^2 + C_3 \bar{p} + C_4 = 0 , \quad (3-4)$$

where

$$C_1 = \frac{2\nu}{n^2(1-\nu)^3} , \quad (3-5)$$

$$C_2 = \frac{1}{(1-\nu)^2} \left\{ 2\nu \bar{I}_y + \frac{1}{n^2} \left[1 + n^2 \frac{GJ_\theta}{\bar{E}I_x} + 2\nu \left(n^2 + \frac{GJ_\theta}{\bar{E}I_x} \right) \right] \right\} \quad (3-6)$$

$$C_3 = \frac{1}{(1 - \nu)} \left[(n^2 - 1)^2 \frac{GJ_\theta}{n^2 \bar{E} I_x} + \left(\bar{I}_y - \bar{I}_{xy}^2 \right) (1 + 2\nu n^2) + \bar{I}_y \frac{GJ_\theta}{\bar{E} I_x} (n^2 + 2\nu) \right], \quad (3-7)$$

$$C_4 = \left(\bar{I}_y - \bar{I}_{xy}^2 \right) (n^2 - 1)^2 \frac{GJ_\theta}{\bar{E} I_x}, \quad (3-8)$$

and,

$$\bar{E} = \lambda + 2G, \quad (3-9)$$

$$\lambda = \frac{\nu E}{(1 + \nu)(1 - 2\nu)}, \quad (3-10)$$

$$\bar{p} = \frac{u_o a r_c}{I_x}, \quad (3-11)$$

$$p_{crit} = \frac{\bar{p} \bar{E} I_x}{r_c^3}, \quad (3-12)$$

$$\bar{I}_{xy} = \frac{I_{xy}}{I_x}, \quad (3-13)$$

$$\bar{I}_y = \frac{I_y}{I_x}. \quad (3-14)$$

The lowest buckling mode, $n=2$, was used as a basis for design calculations.

Torsion of a Composite Section

The formula for tilt waviness shows that that torsional stiffness GJ_θ is the most important stiffness parameters at high values of n ($n=9$ for the present case).

$$\phi = \frac{m_{\theta_o} r_c^2 (1 + An^2)}{EI_x (n^2 - 1)^2} \rightarrow \frac{m_{\theta_o} r_c^2 n^2}{GJ_\theta (n^2 - 1)^2} \text{ for large } n .$$

One expects that the effect of the carbon will be to stiffen the section of interest torsionally in spite of its low shear modulus. To evaluate this case, some new theory was developed for the composite cross section case.

Referring to Figure 3-4, Timoshenko [31] reasons that pure torsion causes a rotational displacement where the displacements are given by

$$u = -\theta zy , \quad v = \theta zx \quad (3-16)$$

where θ is the rate of twist. It can be reasoned the same form of displacement applies to a composite section as shown having zero slip at the boundaries. Warping of the cross section is defined by the use of a warping function ψ so that

$$w = \theta \psi(x,y) \quad (3-17)$$

and this same equation applies. Then, by definition, since normal stresses are assumed to be negligibly small in the torsion problem,

$$\gamma_{xz} = \frac{\partial w}{\partial x} + \frac{\partial u}{\partial z} = \theta \left(\frac{\partial \psi}{\partial x} - y \right) , \quad (3-18)$$

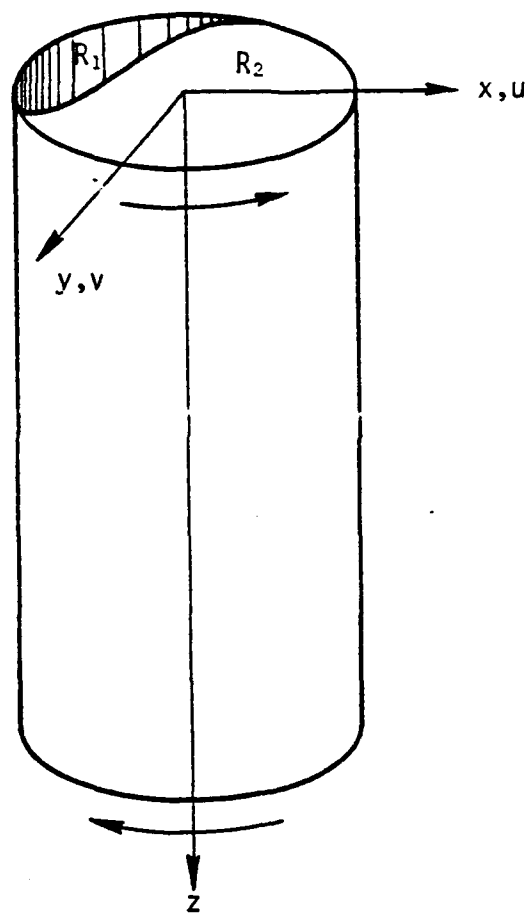


Figure 3-4. Torsion of a Composite Section.

$$\gamma_{yz} = \frac{\partial w}{\partial y} + \frac{\partial v}{\partial z} = \theta \left(\frac{\partial \psi}{\partial y} + x \right) \quad (3-19)$$

and using stress strain relationships

$$\tau_{xz} = G\theta \left(\frac{\partial \psi}{\partial x} - y \right), \quad (3-20)$$

$$\tau_{yz} = G\psi \left(\frac{\partial \psi}{\partial y} + x \right). \quad (3-21)$$

Equations (3-18) through (3-21) are valid for R_1 and R_2 separately.

Using equilibrium and equations (3-20 and 3-21)

$$\frac{\partial^2 \psi}{\partial x^2} + \frac{\partial^2 \psi}{\partial y^2} = 0 \quad (3-22)$$

within each region.

Referring to Figure 3-5 for the development of the boundary conditions and noting the clockwise convention for moving around the section, we have for the shear stress acting normal to a boundary:

$$\tau_n = \tau_{yz} \cos \alpha + \tau_{xz} \sin \alpha \quad (3-23)$$

where

$$\sin \alpha = -\frac{\Delta y}{\Delta s}, \quad \cos \alpha = \frac{\Delta x}{-\Delta s}. \quad (3-24)$$

so

$$\tau_n = -\tau_{yz} \frac{dx}{ds} + \tau_{xz} \frac{dy}{ds} \quad (3-25)$$

or substituting Equation (3-20 and 3-21), we have

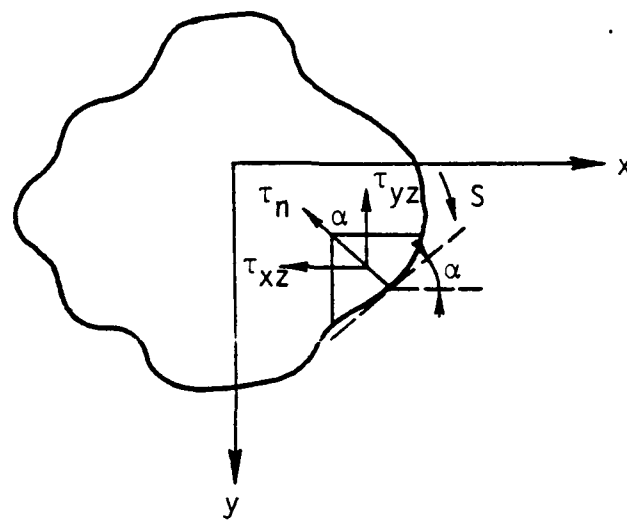


Figure 3-5. Normal Shear Stress Component.

$$\tau_n = G\theta \left(\frac{\partial \psi}{\partial y} + x \right) \frac{dx}{ds} + G\theta \left(\frac{\partial \psi}{\partial x} - y \right) \frac{dy}{ds} . \quad (3-26)$$

On a free boundary

$$\tau_n = 0 \quad (3-27)$$

and on a common boundary between regions

$$\tau_{n1} = G_1\theta \left(\frac{\partial \psi_1}{\partial y} + x \right) \frac{dx}{ds} + G_1\theta \left(\frac{\partial \psi_1}{\partial x} - y \right) \frac{dy}{ds} , \quad (3-28)$$

$$\tau_{n2} = G_2\theta \left(\frac{\partial \psi_2}{\partial y} + x \right) \frac{dx}{ds} + G_2\theta \left(\frac{\partial \psi_2}{\partial x} - y \right) \frac{dy}{ds} . \quad (3-29)$$

The normal shear stresses must transfer across the boundary. Therefore, since θ is the same for both regions,

$$\begin{aligned} & \left\{ G_1 \left(\frac{\partial \psi_1}{\partial y} + x \right) - G_2 \left(\frac{\partial \psi_2}{\partial y} + x \right) \right\} \frac{dx}{ds} \\ & + \left\{ G_1 \left(\frac{\partial \psi_1}{\partial x} - y \right) + G_2 \left(\frac{\partial \psi_2}{\partial x} - y \right) \right\} \frac{dy}{ds} = 0 . \end{aligned} \quad (3-30)$$

So the problem is reduced to finding a function ψ that satisfies both Equation (3-22) and the boundary conditions of Equation (3-30) and (3-27).

Once ψ is found the torsional stiffness is found by taking the moment of the shear stress. This gives the equivalent stiffness as

$$GJ_{\theta} = \sum_{i=1}^n \left(\iint_{R_i} G_i \left(x \frac{\partial \psi_i}{\partial y} + x^2 \right) dx dy - \iint_{R_i} G_i \left(y \frac{\partial \psi_i}{\partial x} - y^2 \right) dy dx \right), \quad (3-31)$$

where the summation is over the n regions. Appendix A contains the finite difference solution to this problem and some examples.

The above method was incorporated into the seal design program so that the proper stiffness GJ_{θ} would be used to predict waviness. Table 3-1 shows the effect of the carbon composite on the stiffness for one of the seal ring designs very close in size to that shown in Figure 3-3. The actual torsional stiffness is four times larger than that based on the metal alone.

After repeated attempts to vary proportions on this design it was concluded that no satisfactory composite section could be designed. The main problem was that for designs where torsional stiffness was low enough such that waviness could be applied, the metal part engaging the secondary seal (Figure 3-2) was too thin to transmit the load from the pistons to the ring without bending. Or, in many otherwise satisfactory designs, the required forces were larger than available. The problem comes about because making the seal long enough to provide moment arm space causes the stiffness to be too high for the forces available when the part is made of metal.

TABLE 3-1

Composite Section Stiffness

	\bar{x}	\bar{y}	$G^* J_{\theta}$	Moment Arm Needed
Metal Only	.2184 in	.2790 in	1446 lb in ²	.1680 in
Composite Section	.2193	.2795	4242	.4386

All Carbon Seal Design

The solution to the above problem is to make the entire ring out of low modulus carbon. Thus the ring does not get too stiff for the size needed to accommodate the force application.

Figure 3-6 shows the configuration of the all carbon seal design. It is essentially the same shape as before. This change not only reduces the stiffness, and therefore the needed moment arm, but also eases design, fabrication, and assembly problems. The design computer program for the previous configuration was modified slightly to accommodate this all carbon configuration and design proceeded accordingly.

Warping Analysis

One additional area that must be considered is the effect of warping on the stiffness of the cross section. Warping is the nonuniform z direction displacement which accompanies torsion. Oden [32] shows that warping can induce stresses in the section that will have the effect of increasing the torsional stiffness. This occurs when warping is constrained as it is in the alternating torsion of a wavy ring.

Equations for waviness where warping is significant will now be derived. Warping makes the following change in the stress resultant-displacement equations:

$$M_{\theta} = \frac{GJ}{R_c} (\phi' + \bar{v}') - \frac{E\Gamma^*}{R_c^3} (\phi''' + \bar{v}''') , \quad (3-32)$$

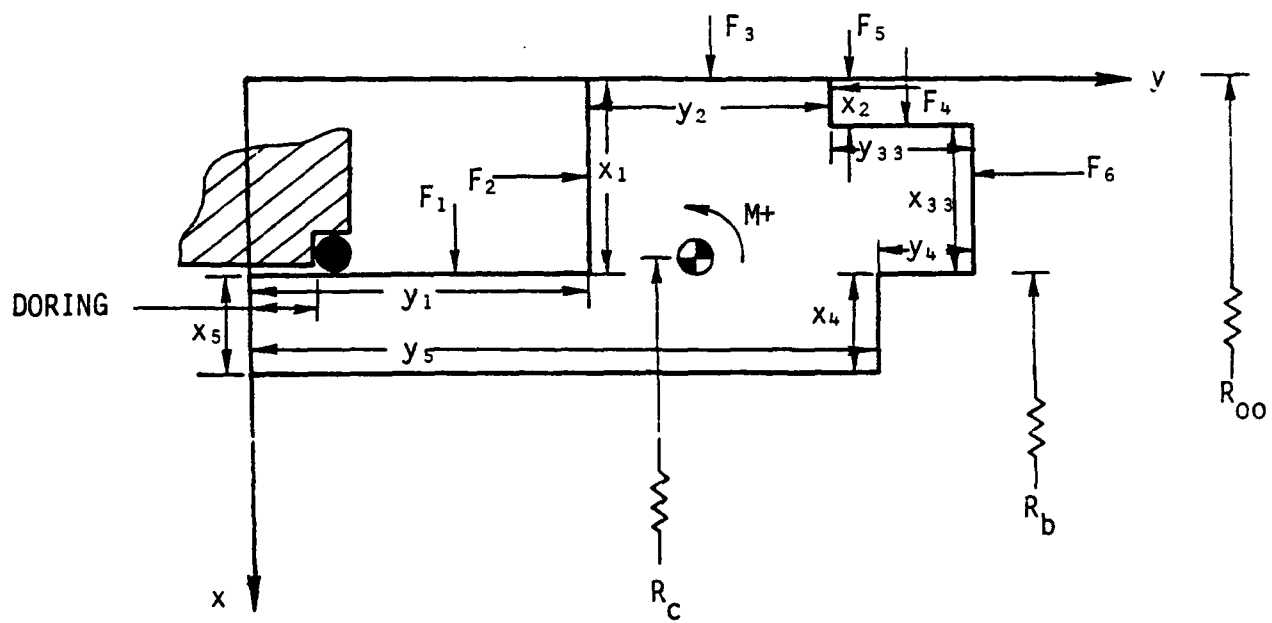


Figure 3-6. A11 Carbon Seal Design

where Γ^* is the warping constant and all other ring equations remain the same [33].

The equilibrium equations are [34]

$$M_\theta - M_x + m_\theta R_c = 0 , \quad (3-33)$$

$$M_x'' + M_\theta' = 0 . \quad (3-34)$$

Using the previous equation for M_x and M_θ from [33] gives:

$$(\phi'' - \bar{v}''''') + \frac{1}{A} (\phi'' + \bar{v}'') - \frac{1}{B} (\phi'''' + \bar{v}''''') = 0 \quad (3-35)$$

and

$$\frac{1}{A} (\phi'' + \bar{v}''''') - \frac{1}{B} (\phi'''' + \bar{v}''''') - (\phi - \bar{v}'') + \frac{m_\theta R_c^2}{EJ_x} = 0 , \quad (3-36)$$

where

$$A = \frac{EJ_x}{GJ_\theta} \quad \text{and} \quad B = \frac{J_x R_c^2}{\Gamma^*} . \quad (3-37)$$

Assuming that

$$m_\theta = m_{\theta 0} \cos n\theta \quad (3-38)$$

it follows that

$$\phi = \phi_0 \cos n\theta \quad \text{and} \quad \bar{v} = \bar{v}_0 \cos n\theta . \quad (3-39)$$

Substitution gives

$$\phi_o = \frac{m_{\theta_o} R_c^2}{EJ_x} \frac{1}{\left(\frac{n^2}{A} + \frac{n^4}{B} + 1\right) - \frac{\left(1 + \frac{1}{A} + \frac{n^2}{B}\right)}{\left(n^2 + \frac{1}{A} + \frac{n^2}{B}\right)} \left\{\frac{n^2}{A} + \frac{n^4}{B} + n^2\right\}} \quad (3-40)$$

and

$$v_o = \frac{m_{\theta_o} R_c^3}{EJ_x} \frac{1}{\left(\frac{n^2}{A} + \frac{n^4}{B} + 1\right) - \frac{\left(n^2 + \frac{1}{A} + \frac{n^2}{B}\right)}{\left(1 + \frac{1}{A} + \frac{n^2}{B}\right)} \left\{\frac{n^2}{A} + \frac{n^4}{B} + n^2\right\}} \quad (3-41)$$

Equations (3-40) and (3-41) are the modified form used in the design program to predict waviness considering warping.

It now becomes necessary to determine Γ^* for the particular cross section in question. This was accomplished by again using a finite element program and using three-dimensional solid elements (eight node brick). The section considered was a bar of exact cross section and length equal to that of one half wave (of nine waves) around the seal (see Figure 3-7). A moment was applied to the end of the bar and the resultant twist was calculated. In this calculation the z displacements at both ends were not constrained, only x and y. Next, the z displacements were constrained and the angles of twist again calculated. This simulates the symmetry conditions where warping must go to

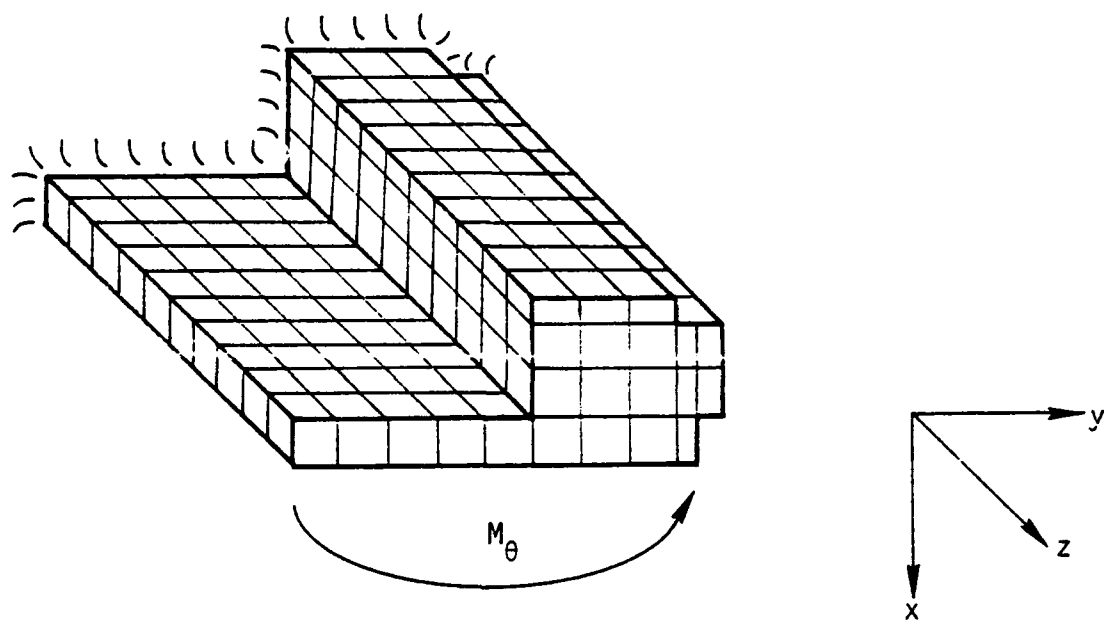


Figure 3-7. 3-D Finite Element Configuration.

zero in the alternating twist. Table 3-2 shows the results for a cross section very close in dimensions to the final solid carbon design.

Now the straight bar version of Equation (3-32) is:

$$EI^*\phi'''' - GJ_{\theta} \phi' + M_{\theta} = 0 . \quad (3-42)$$

For M_{θ} constant solving for ϕ gives

$$\phi = C_1 e^{rx} + C_2 e^{-rx} + \frac{M_{\theta} x}{GJ_{\theta}} + C_3 , \quad (3-43)$$

where

$$r = \sqrt{\frac{GJ_{\theta}}{EI^*}} . \quad (3-44)$$

The boundary conditions are

$$\begin{aligned} \phi(0) &= 0 , \\ \phi'(0) &= 0 , \\ \phi'(l) &= 0 , \end{aligned} \quad (3-45)$$

where $\phi' = 0$ again simulates the symmetry conditions. The final solution is

$$\begin{aligned} \phi = & - \frac{M_{\theta}}{rGJ_{\theta}} \left[\frac{(e^{-rl} - 1)}{(1 - e^{-2rl})} + 1 \right] e^{rx} - \frac{M_{\theta}}{rGJ_{\theta}} \frac{(e^{-rl} - 1)}{(1 - e^{-2rl})} e^{-rx} \\ & + \frac{M_{\theta} x}{GJ_{\theta}} + \frac{M_{\theta}}{rGJ_{\theta}} \left[\frac{2(e^{-rl} - 1)}{(1 - e^{-2rl})} + 1 \right] . \end{aligned} \quad (3-46)$$

Dividing through by $M_{\theta} l / G_{\theta}$ and evaluating at $x = l$ gives,

TABLE 3-2

Warping Function Calculation

	twist (in/in)	$G J_{\theta}$ (lb-in ²)
Z Deflections Unconstrained	.002703	3575
Z Deflections Constrained	.001742	--

$$\frac{\phi(\lambda)}{\frac{M_{\theta}\lambda}{GJ_{\theta}}} = -\frac{1}{r\lambda} \left(\frac{(e^{-r\lambda} - 1)}{(1 - e^{-2r\lambda})} + 1 \right) e^{r\lambda} - \frac{1}{r\lambda} \left(\frac{(e^{-r\lambda} - 1)}{(1 - e^{-2r\lambda})} \right) e^{-r\lambda} + 1$$

$$+ \frac{1}{r\lambda} \left[\frac{2(e^{-r\lambda} - 1)}{(1 - e^{-2r\lambda})} + 1 \right]. \quad (3-47)$$

The left-hand side is the ratio of twist for the condition of the z displacements constrained to the condition of the z displacements unconstrained. Going back to Table 3-2 and using those values we have for the problem of interest

$$\frac{\phi(\lambda)}{\frac{M_{\theta}\lambda}{GJ_{\theta}}} = 0.644596. \quad (3-48)$$

Now, letting $\lambda = 0.543$ in., the length of the bar section, and using a root finding technique to solve Equation (3-47) for r for the condition of Equation (3-48) we get

$$r = 10.286/\text{in.} \quad (3-49)$$

Then, using Equation (3-44) and section properties determined later we have

$$\Gamma^* = 1.0144 \times 10^{-5} \text{ in.}^6 \quad (3-50)$$

and Equation (3-37) gives

$$B = 3097. \quad (3-51)$$

Design Solution

Using the above information in the design program and the criteria discussed, final satisfactory design dimensions were found and are shown in Table 3-3.

Predicted Performance

Previously developed computer programs [5] were used to predict the performance of the new design. Figures 3-8 and 3-9 show these results. The friction torque is high at lower speeds but drops very quickly for increasing rpm. Even at zero speed the friction is less than half as much as would be expected with nowaviness. The decreased torque at higher speed is due to hydrodynamic effects which result in 100 percent fluid pressure load support at the higher speed.

Figure 3-9 shows that leakage is quite small, less than 1.5 cc/min up to 1800 rpm for all cases. A slightly larger value of ϕ_0 was used for this design compared to the first design (400 as compared to 320) so that sufficient wave would be available for hydrodynamic effects. Leakage is increased by about one-half a cubic centimeter per minute at the worst case compared to the previous design.

Figure 3-10 shows the comparison of the worn profile shapes for the extremes of operating conditions. As can be seen, the profile changes very little and as a result additional wear is minimized as the operating conditions are changed. This is as expected using nine waves.

TABLE 3-3

Solid Carbon Seal Design
(Ref. Fig. 3-6)

ϕ_o	$= 400 \times 10^{-6} \text{ in/in}$	
d	$= .316 \text{ in}$	
x_1	$= .2345 \text{ in}$	$y_1 = .4200 \text{ in}$
x_2	$= .0470 \text{ in}$	$y_2 = .2600 \text{ in}$
x_3	$= .1875 \text{ in}$	$y_3 = .2100 \text{ in}$
x_4	$= .0900 \text{ in}$	$y_4 = .0600 \text{ in}$
x_5	$= .0900 \text{ in}$	$y_5 = .8300 \text{ in}$
R_{pi}	$= 1.8100 \text{ in}$	
R_b	$= 1.900 \text{ in}$	
R_{oo}	$= 2.1345 \text{ in}$	
R_c	$= 1.9427 \text{ in}$	
DORING	$= .2026 \text{ in}$	
\bar{x}	$= .1918 \text{ in}$	
\bar{y}	$= .5453 \text{ in}$	
I_x	$= .0083085 \text{ in}^4$	
I_y	$= .0014610 \text{ in}^4$	
$G J_\theta$	$= 3183 \text{ lb}\cdot\text{in}^2$	
J_θ	$= .002546 \text{ lb}\cdot\text{in}^2$	
A	$= 7.83$	
v	$= 5.687 \times 10^{-6} \text{ in}$	
e_{cc}	$= .2053 \text{ in}$	
Critical buckling pressure = 1007 lb/in		
Pressure caused rotation = $1.1 \times 10^{-9} \text{ in/in}$		

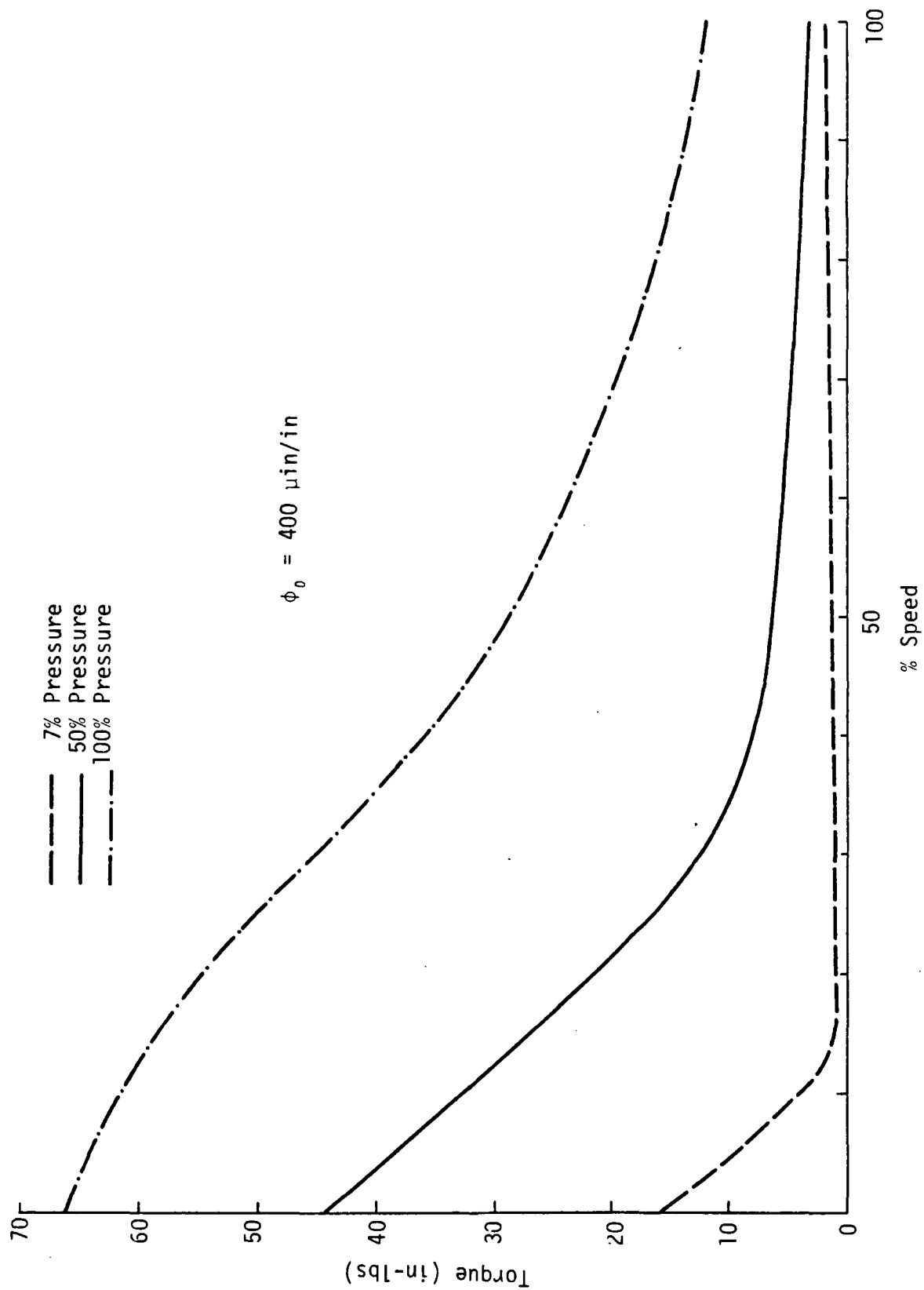


Figure 3-8. Predicted Torque

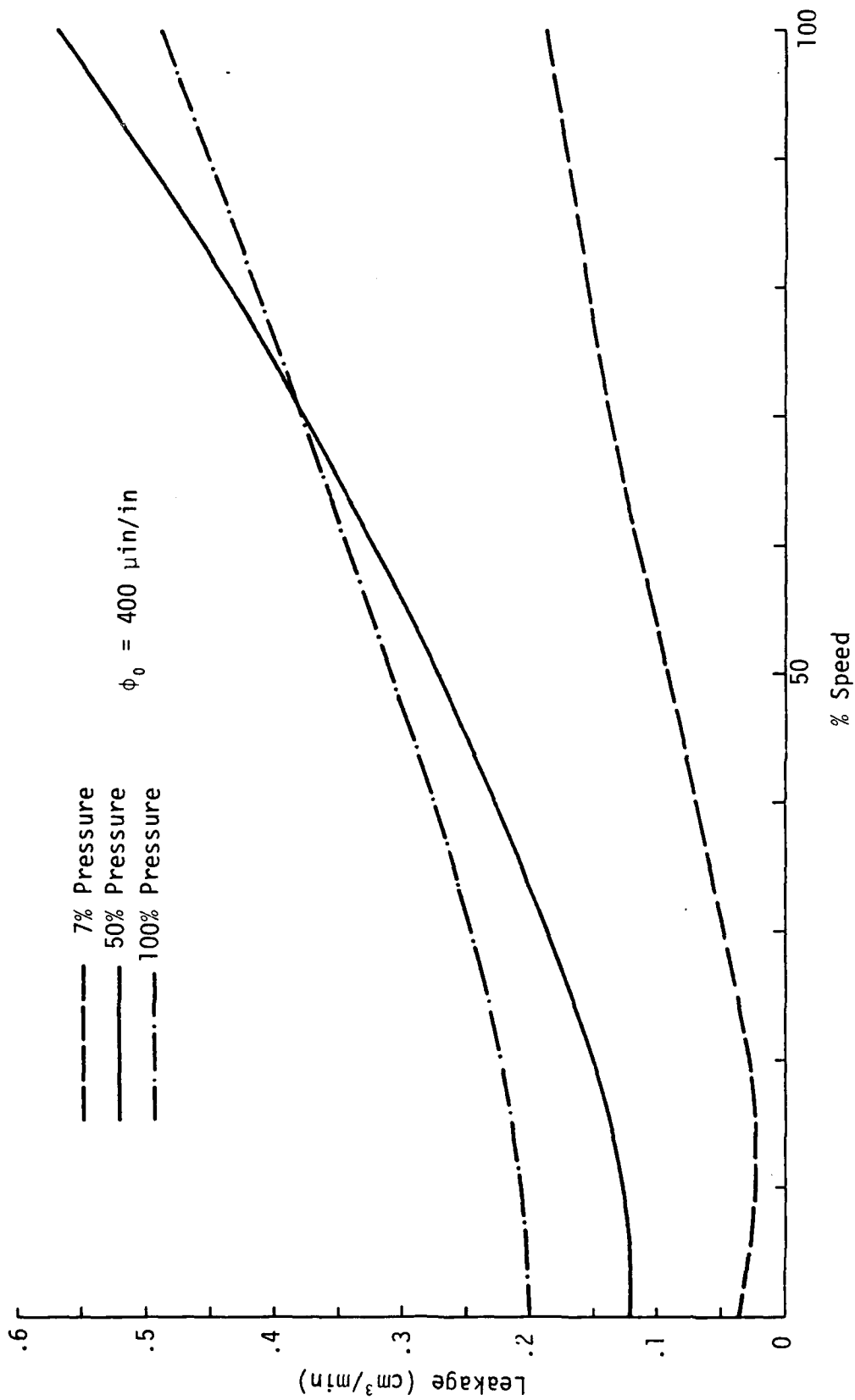


Figure 3-9. Predicted Leakage.

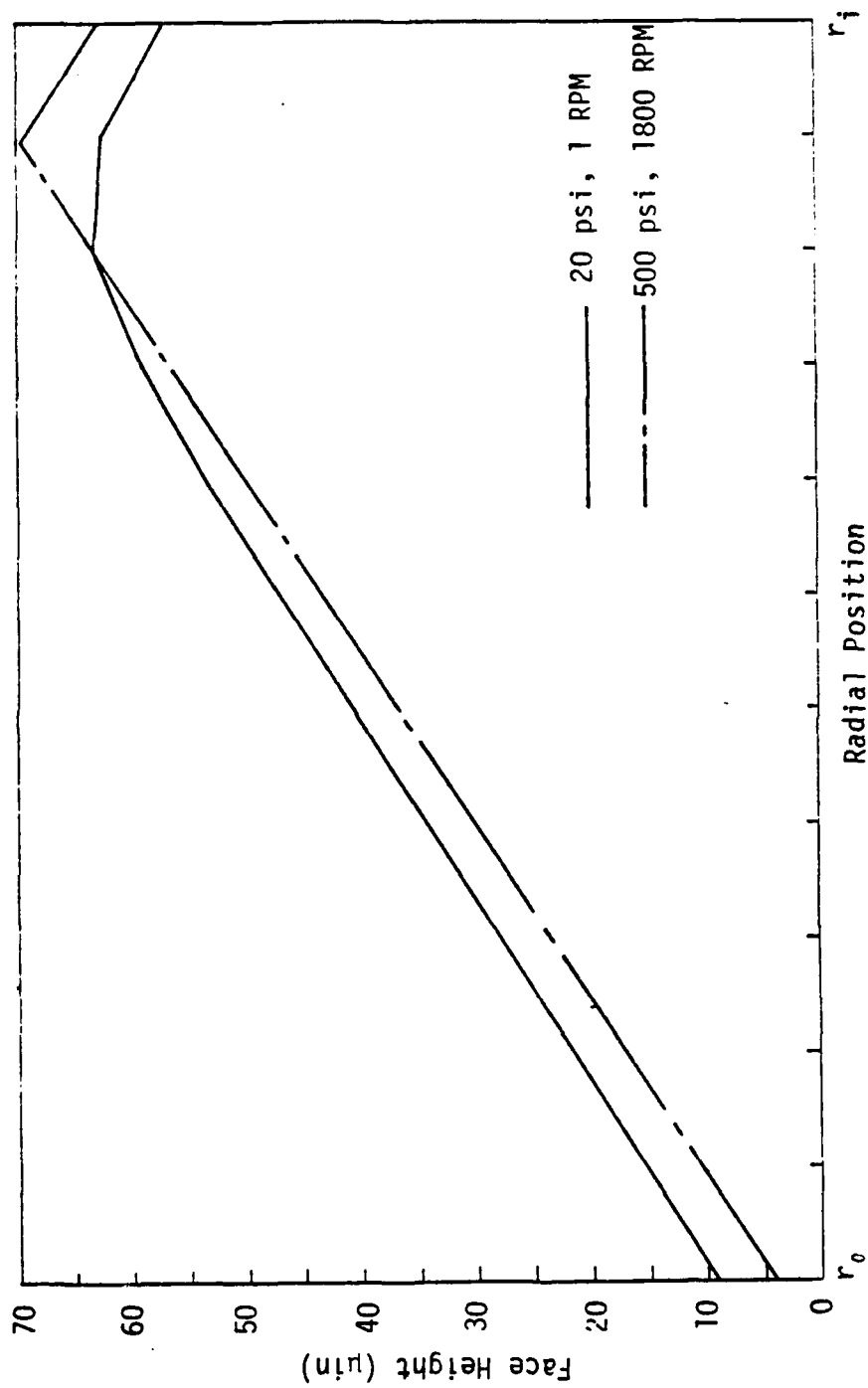


Figure 3-10. Predicted Worn Face Shape.

Strength Analysis and Test

Because of the reduced cross section of this all carbon design, a primary concern becomes the strength of the carbon in relation to the waviness loads applied as shown in Figure 3-11. Each piston produces a maximum load of 86 lb distributed as shown but of course at discrete intervals around the seal ring. The carbon was modeled by finite elements as shown using an assumption of axisymmetric loading to get a first approximation. It was found that the maximum tensile stress is about 4000 psi on element No. 8 in Figure 3-11. The tensile strength is only 8000 psi.

Given that the approximate solution was not considered conservative because the concentrated loads had to be distributed to obtain the axisymmetry needed for solution, other methods to calculate the maximum stress were tried but none were considered accurate enough to be relied upon. The problem is clearly a 3-D stress analysis problem, the modeling for which is very cumbersome. Thus, it was decided that the only reliable way to assess the adequacy of the strength of the part was by actual test.

Figure 3-12 shows a cross section of the test set up. The carbon shown was machined out of an existing carbon ring to similar dimensions and the same cross sectional area as the new design. Pressurized oil was slowly introduced into the test fixture and the pressure monitored by computer and pressure transducer. Failure was localized at the point of application and occurred at a load of 177 lb. The maximum

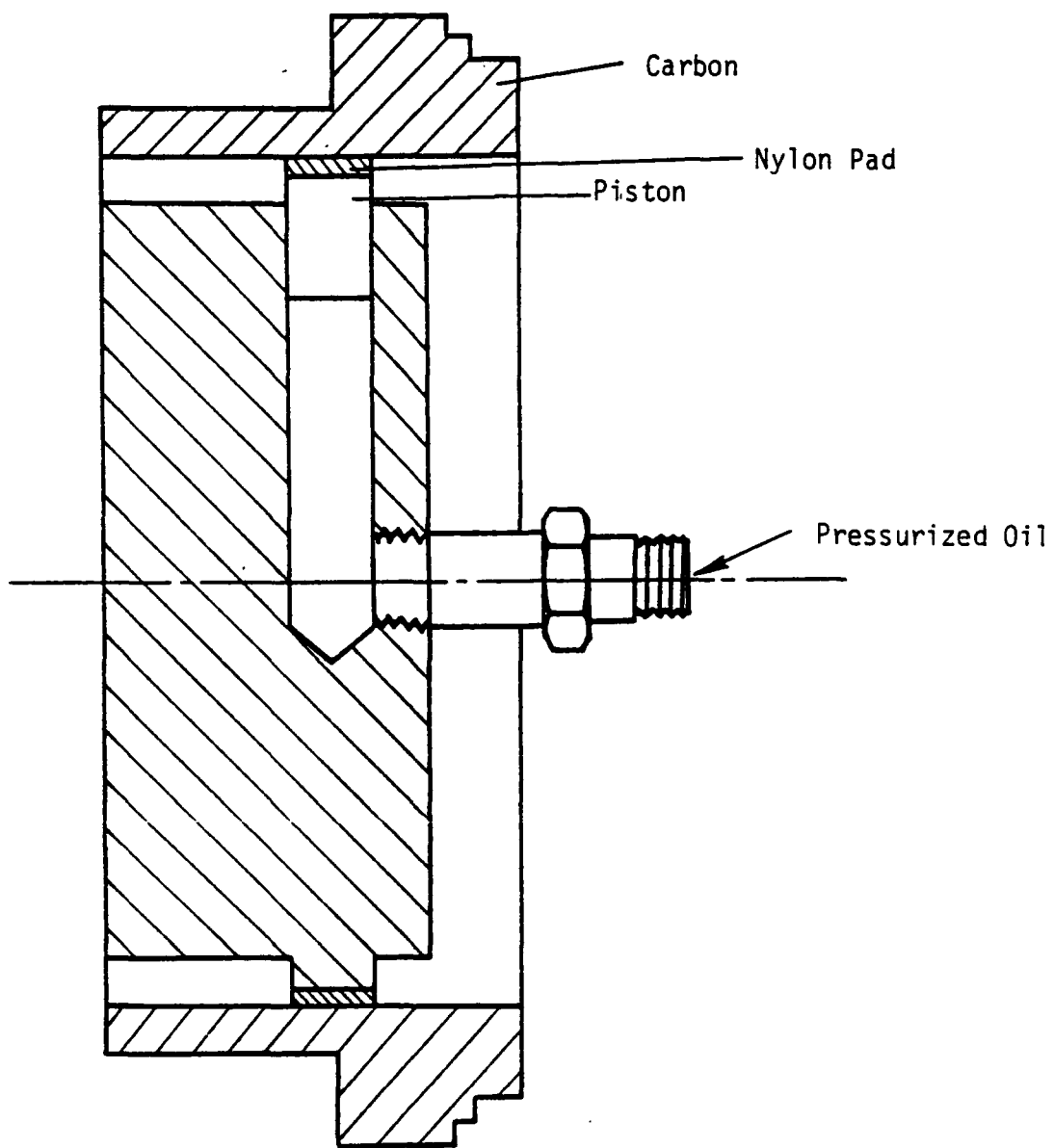


Figure 3-12. Destructive Test Set-up.

design piston load, as stated before is 86 lb. This indicates a safety factor of 2.05.

The question of fatigue strength also had to be addressed. Paxton [35] states that the terminal fatigue strength of most carbon is thought to be between 45 and 85 percent of the original tensile strength. Thus for the present design, the safety factor could be as low as 1.0 but most likely higher. It was decided since no specific fatigue data was available, the design would be considered acceptable on a prototype basis.

Other Calculations

Many other design calculations were made but will not be presented in detail here. However, the subject areas were:

- 1) O-ring friction on spring seat.
- 2) Stresses in waviness inducer.
- 3) Piston stresses.
- 4) Pressure caused rotation of spring seat.
- 5) Buckling of spring seat.
- 6) Hoop stress caused by piston load and band clamp stress for split design.

Concerning the last item, it was decided early on that the carbon had to be clamped at its OD with a metal band so as to place a large enough compressive stress on the carbon to overcome the tensile stress caused by the piston loads. This would allow the design to be split in half and simply be clamped together.

Final Design

Figure 3-13 shows the assembly view of the seal. Starting at the left, one of three sinusoidally varying pressures is directed through two (180° apart) pressure channels in the waviness cylinder (1). It was desired that the existing waviness cylinder be used since it already had the necessary tilt and offset machined into it. This waviness cylinder was therefore modified so that it could be used with the present design. One of the modifications was the addition of three drilled and tapped holes (2) (one of which is shown). These intersect the pressure channels in the waviness cylinder. The pressure is then directed through one of three swagelok fittings (14) to one of three 1/16 in. monel inlet tubes (13) which delivers the pressure to the waviness inducer (9) and terminating at a set of 18 pressure pistons (11) via connecting tubing (12). The pressure induces a force through the pressure piston (11) to a delrin spacer (10) which causes a counter-clockwise moment (in this particular cross-section) to the all-carbon seal (7). The seal is a pure carbon P658RC carbon. The seal is driven by the drive ring (5) through the drive ring adapter (6). This arrangement allows for the carbon to "float" and therefore take its alignment from the face of the rotating secondary seat (8) which is a carborundum KT[®] silicon carbide. Both primary and secondary rings are of zero moment design, i.e., no rotation of the rings under water pressure. Preload is provided by the springs housed in the spring retainer (3) through the spring seat (4). The design of the spring seat is such that the secondary O-ring seal (15) moves with the carbon ring with

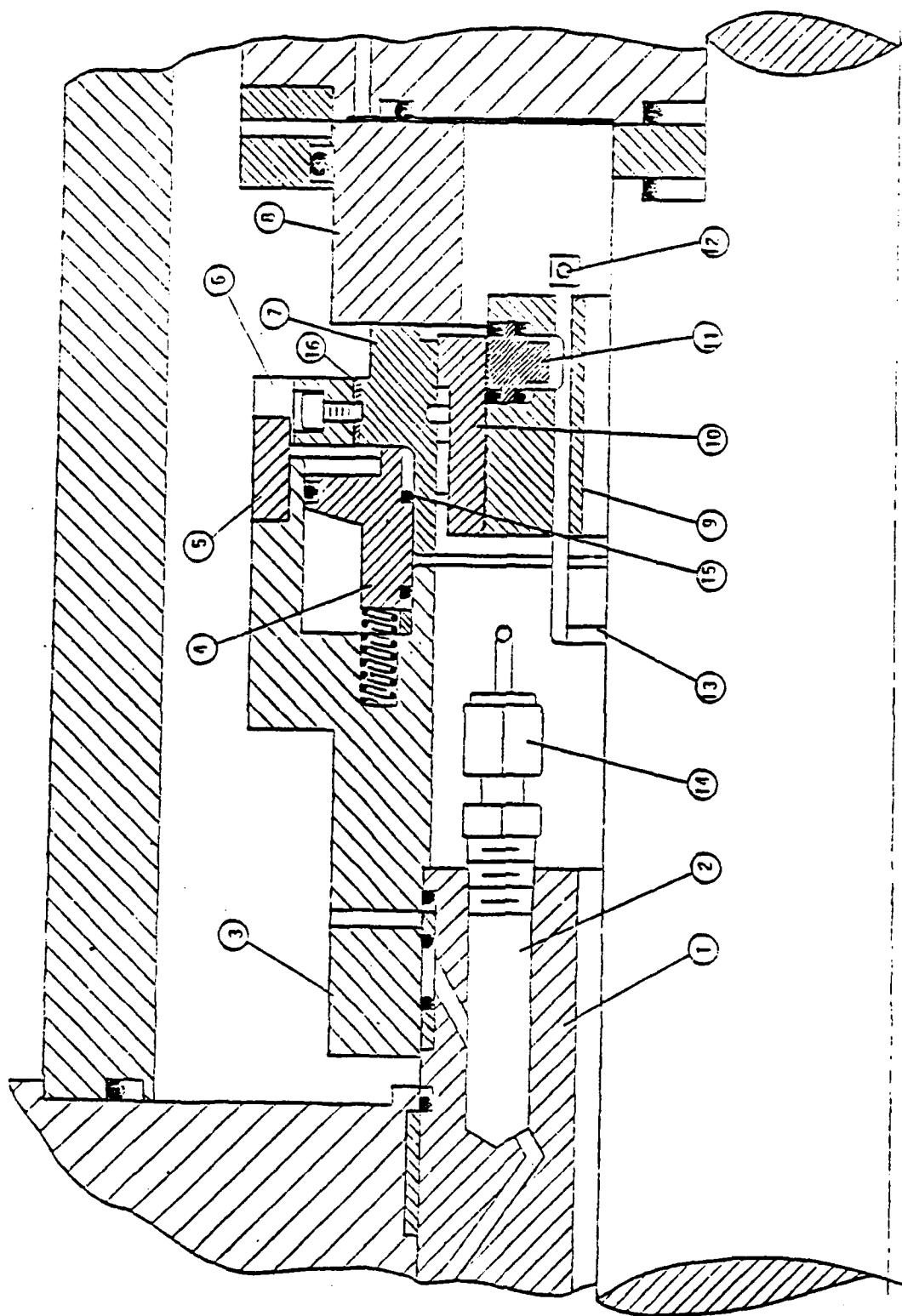


Figure 3-13. Nine-wave Seal Assembly View, Second Design.

wear so the pressure balance is relatively unaffected. The clamp band (16) produces a radial preload to the carbon ring (7) for the case when the carbon will be split.

Waviness of the new seal was measured external to the test rig by the use of the waviness inducer and the addition of a fixture that would hold pressure on 18 of the 54 pistons. Pressurizing 18 pistons gives one complete set of nine waves. The apparatus was placed on the precision rotary table and axial displacement traces near the OD of the carbon face were taken with the stylus of a surface analyzer. The signal was digitized by a computer and a Fourier analysis of waviness components was made. Table 3-4 shows the results of the waviness measurements.

The results show that the first waviness measurements were considerably lower than the 60 μ in. design value. It was determined that the original Delrin band, which transmits the load from the piston of the waviness inducer to the inside diameter of the seal ring, was absorbing the wave by distributing the load over a much larger area than the area of the pistons. This was checked by making 18 individual Delrin pads which fit on the pistons, making each piston independent of any other and concentrating the load at the point of application. Results showed a dramatic increase in the harmonic waviness.

Since the original Delrin band was by design, necessary for waviness inducer-carbon seal alignment, the band was modified. This modification was accomplished by machining 54 pad locations leaving only 0.020 in. of material between pads on the inside diameter. This allowed for more individual freedom of movement for each piston and

Nine-Wave Amplitude Study

Waviness Amplitude Near Seal O.D.

* Equivalent to actual test pressure.

less absorption by the ring as a whole. The waviness results are shown in Table 3-4.

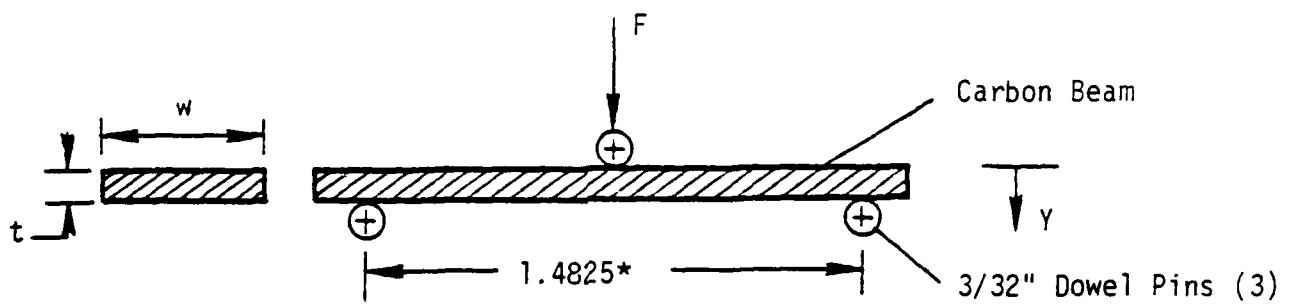
Secondly it was found that the metal clamp band greatly stiffened the seal ring torsionally (Table 3-4). For the purpose of these and subsequent tests it was removed with the question of how to put it in place for the split ring without stiffening the ring being left unanswered.

After these two modifications were made, the measured waviness started to approach the calculated waviness although it was still considered somewhat low. Further explanations were sought.

Young's Modulus Tests

An experiment was set up to check the value of Young's modulus for P658RC carbon. The test apparatus is shown in Figure 3-14. Three carbon beams were machined out of a carbon seal ring and then ground to size. The beams were simply supported by two 3/32 in. diameter dowel pins. The load was applied at midpoint through another 3/32 in. diameter dowel pin and the resultant deflection measured by a 0.0001 in. indicator at that point. Simple beam theory was then used to calculate the value for E. The results are given in Table 3-5.

From these results it was concluded that the value of $E = 3.0 \cdot 10^6$ psi used in design was reasonably close to the measured result. It was decided that an ample wave was available for test.



*measured by profilometer.

Figure 3-14. Test Setup for Young's Modulus Experiments.

TABLE 3-5

YOUNG'S MODULUS RESULTS

FOR P658RC CARBON

Beam #	w (in)	t (in)	I (in ⁴)	F/y (lb/in)	E (lb/in ²)
1	.3824	.0602	6.9522×10^{-6}	327.2	3.19×10^6
2	.3825	.0602	6.9540×10^{-6}	323.3	3.15×10^6
3	.3826	.0602	6.9559×10^{-6}	323.9	3.16×10^6

2000 Hour Test Results

Performance During Test

Figures 3-15 and 3-16 show the performance of the seal one week after start and then eight weeks later. Each plot is for one complete weekly cycle. These are presented so as to compare the initial and final operation. The performance is much the same on the average. The torque fluctuations are somewhat higher, at the 14 percent speed and 100 percent pressure conditions, at the beginning than at the end of the test. The 100 percent speed and pressure conditions show a torque which is quite the same throughout the test, averaging about 2 N·m both in the forward and reverse directions. The total leakage during the second week of operation (Figure 3-15) was 253 cc, which resulted in an average leakage rate of 0.025 cc/min. Figure 3-16 has a total leakage for the week of 321 cc, which is a leakage rate of 0.032 cc/min. The spikes are the result of leakage getting trapped because of surface tension within the leakage path and then suddenly flowing.

One problem encountered during the test is illustrated by Figure 3-17. The torque readings became erratic as a result of insufficient waviness. This was caused by a clogged oil supply line filter to the waviness generator. The filter was cleaned and replaced at about 412 hours into the test. This eliminated the erratic behavior. This problem occurred a few more times during the test and in each case the filter was replaced with a new one.

One problem occurred during the test. The time of its first occurrence is not exactly known, but it became more pronounced toward the end of the test. Figure 3-18 shows uniform large fluctuations in

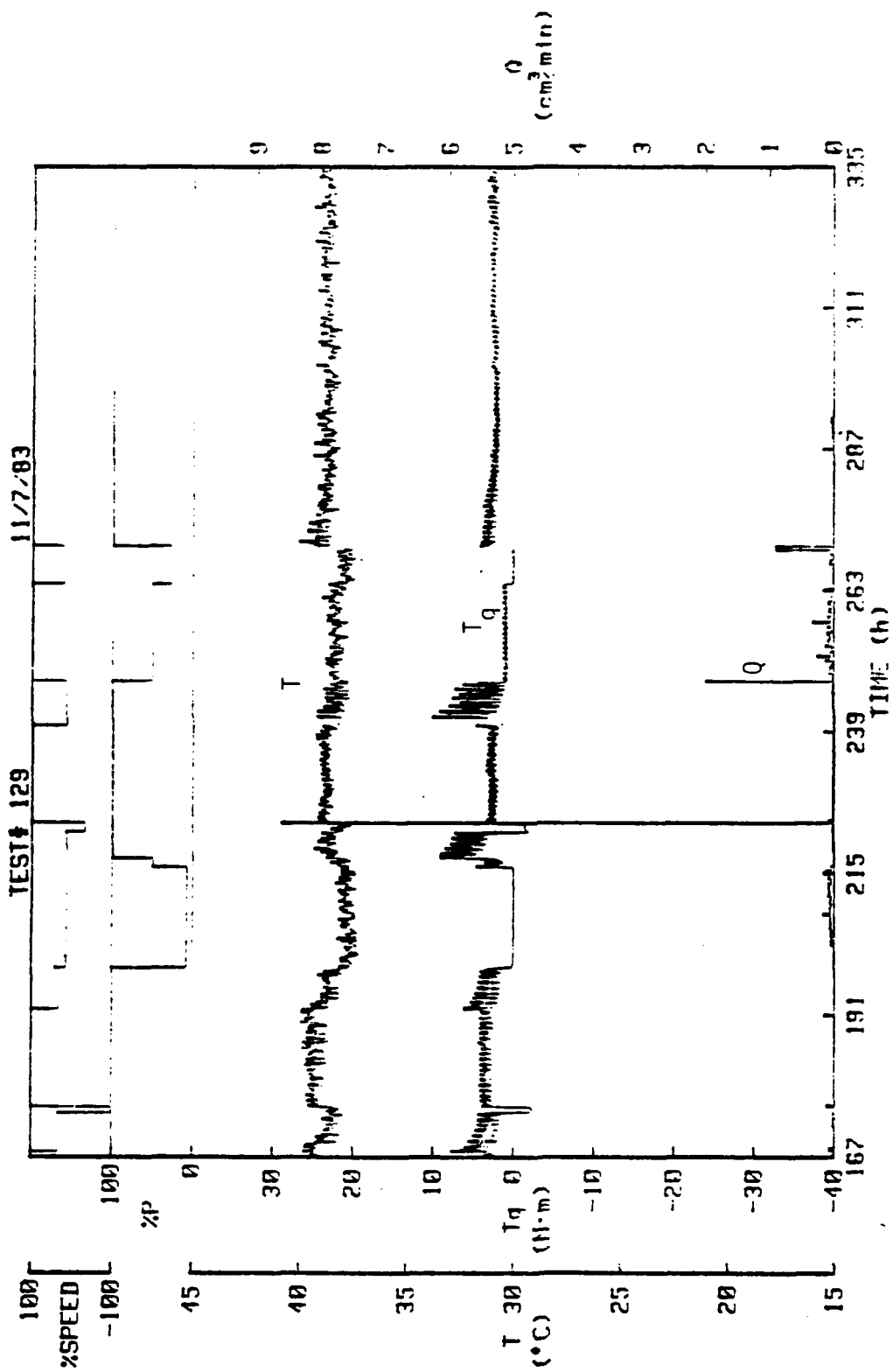


Figure 3-15. Test #129--167-335 Hours.

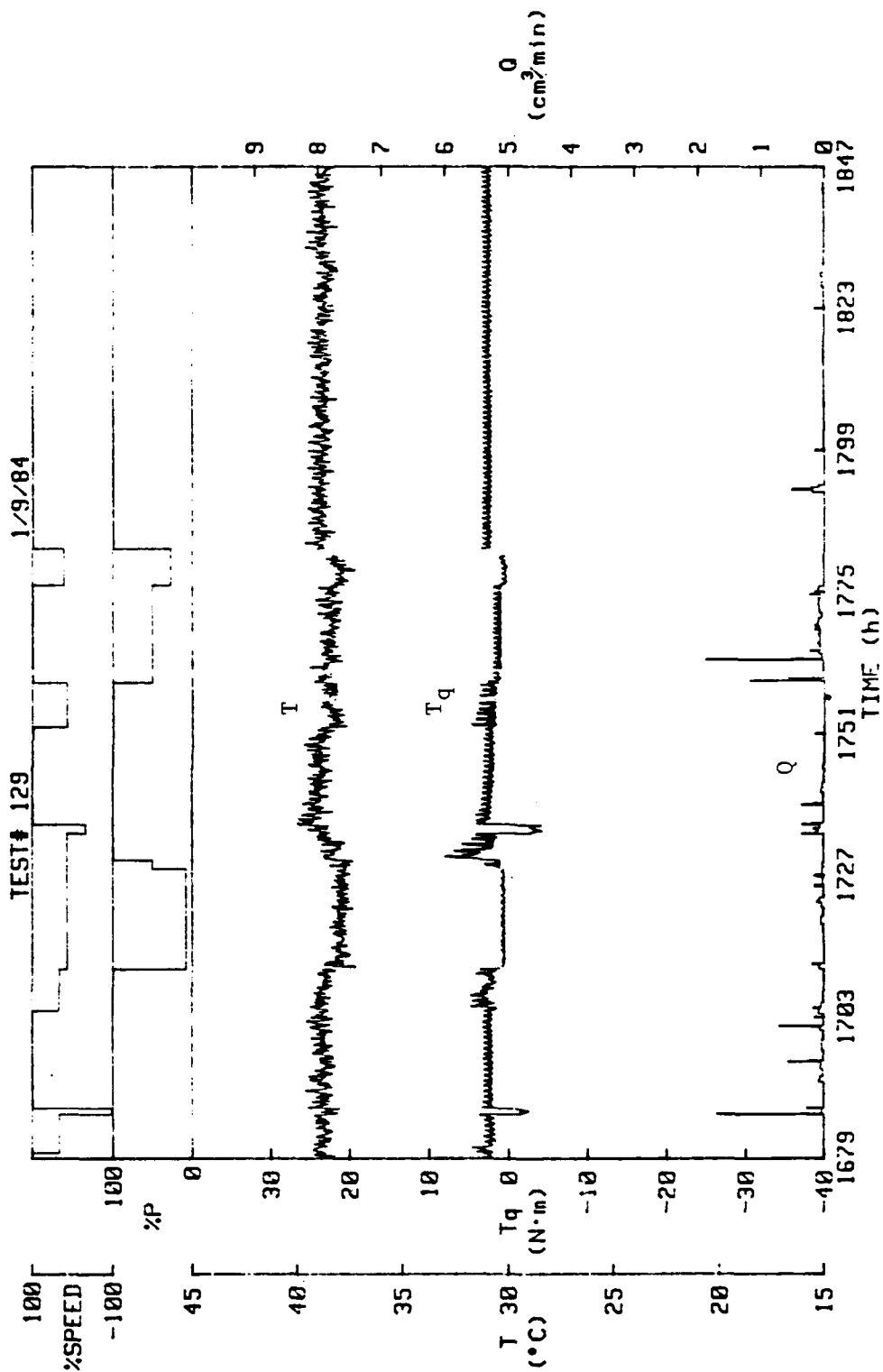


Figure 3-16. Test #129--1679-1847 Hours.

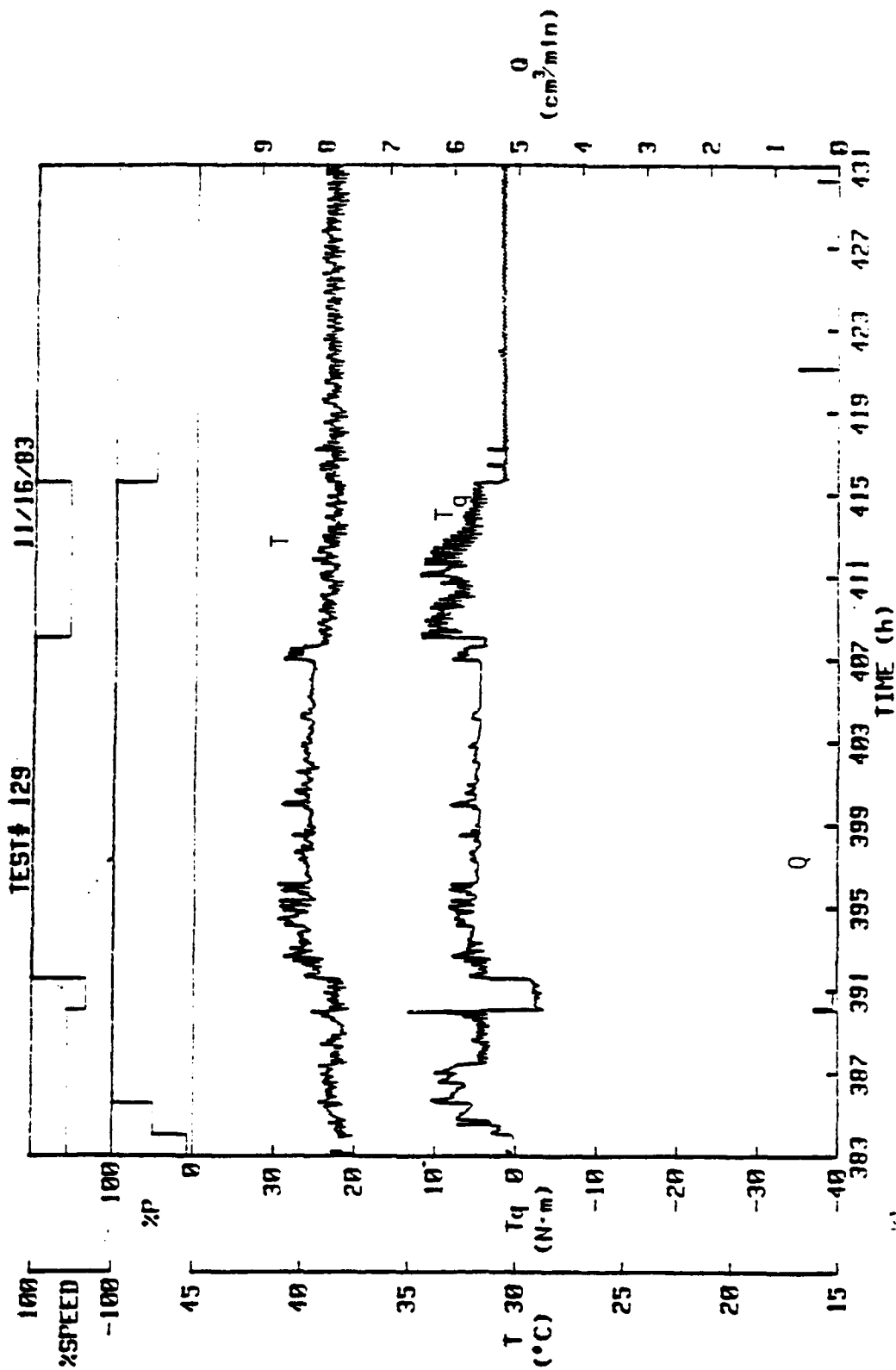


Figure 3-17. Test #129--383-431 Hours.

torque at the different operating conditions. On disassembly it was observed that the seal thermocouple was severely bent due to some relative motion between the carbon seal and the spring retainer. The reason for this angular displacement is due to the fact that the seal slipped in its silicon rubber bond. As a result, some small angular rotation of the carbon resulted causing the thermocouple to bind and ultimately become severely bent. This introduced a component of force which coupled with the movement of the wave around the seal, caused the fluctuations observed in Figure 3-18.

Silicon rubber bonding material was used so as to allow for the differential rates of thermal expansion for the carbon and the monel drive ring and also to be able to withstand the forces needed to drive the seal. The rubber adhered very well to the carbon but the bond was poor with the monel.

Post Test Analysis

Disassembly also showed that five springs behind the spring seat were bent due to the rotation of the carbon as stated above. One of these springs was broken. The epoxy case covering tubes in the waviness inducer had a slight bulge in it indicating some seepage of oil from the tubes inside. Because of the design, however, the oil could not contaminate the sealed fluid since the waviness mechanism is on the zero pressure side.

Radial profiles of the carbon showed a divergent taper of $-763 \mu\text{m/m}$ average, this being the result of pressure caused rotation of the seal. Table 3-6 shows the wear results. The average wear for the 2000

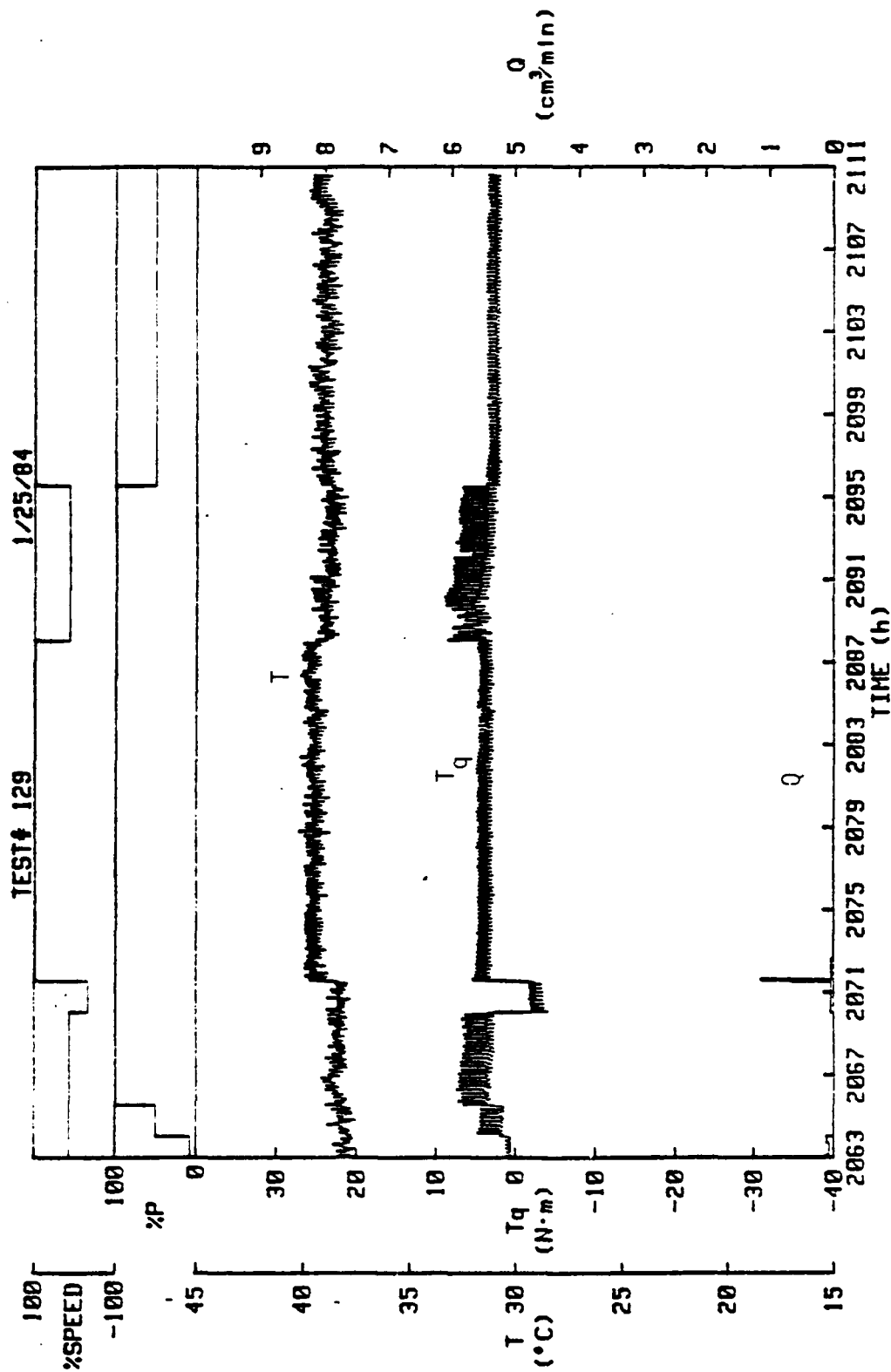


Figure 3-18. Test #129--2063-2111 Hours.

TABLE 3-6
Wear Results (2000 hours)

Position	Initial Wear Groove Depth (μ in)	Final Wear Groove Depth (μ in)	Wear (μ in)
1	2800	2500	300
2	2860	2500	360
3	2240	1880	360
4	2520	2120	400
Average			355

hour test was 355 μ in. Radial traces of the K_T SiC seat showed wear of about 15 to 20 μ in.

Comparison to Theory

Using a numerical model of wavy seal operation developed previously [5], performance was predicted for the various operating conditions in the 2000 hour test as shown in Figures 3-8 and 3-9. For the experimental results, six weeks of operation were used to establish average values at the different operating conditions. These values are given in Table 3-7. The results show a wide variation sometimes even though the apparatus was rezeroed weekly. These results are compared in Table 2-4 for drive torque only.

The table shows that there is a good relationship between predicted and experimental torque, although under some conditions agreement is not as close as desired. Perhaps most importantly is that the reduction of friction with increasing speed shown by the theoretical data is followed by the experimental data although the experiment does not show as strong a relationship. This agreement verifies the predicted operation of hydrodynamic effects in water. In earlier tests, face geometry could change with speed. In this test, the nine waves have a very high stiffness and remain constant and geometry does not change with speed. The other very encouraging result in Table 3-8 is that the speed effects are symmetrical. This indicates that some type of torque caused geometry change is not really influencing the results and more importantly, that the average torque values are reliable because they repeat in the opposite direction.

TABLE 3-7

One Week Average Torque Values
Test 129 (N · m)

<u>Speed %</u>	<u>Week</u>					
	671 to 839	839 1007	1007 1175	1571 1679	1679 1847	2015 2183 hrs
14/7	0.42	0.61	0.77	0.96	0.65	0.70
14/50	2.55	7.93	1.36	1.98	1.58	2.22
14/100	7.19	7.60	2.36	2.95	2.92	4.42
33/100	5.20	3.58	3.25	3.16	2.63	3.58
100/50	1.05	1.23	1.22	1.70	1.28	2.51
100/100	2.91	2.95	2.45	2.66	2.53	4.33
-33/100	-5.05	-4.76	-2.91	-2.51	-3.34	-2.50
-100/100	-2.39	-3.11	-2.80	-5.25	-1.73	-2.33

TABLE 3-8

Comparison of Experimental and
Theoretical Results--Design 2
Torque* (N · m)

Experimental† Theory	% Speed				
	-100	-33	14	33	100
7	** -0.3	** -0.1	.7 .2	** .1	** .3
50	** -0.4	** -1.2	2.9 3.2	** 1.2	1.5 .4
100	-2.9 -1.4	-3.5 -4.7	4.6 6.7	3.6 4.7	3.0 1.4

*Leakage was not compared because of erratic
rates caused by collection passages.

**Conditions not run experimentally.

†Based on six weeks data from Test #129.

With regard to leakage, the model was used to predict a weekly average rate of 0.4 cc/min. The average measured weekly for the week of operation shown in Table 3-9 was 0.028 cc/min. Thus leakage does not agree well at all. The experimental result of course is favorable since it is lower than predicted. However, it is known that the model is weak in predicting leakage. Previous work [16] showed that in contact problems like this one actual leakage is very hard to predict to better than an order of magnitude.

Split Ring Design

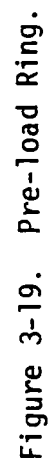
To meet ultimate submarine needs, it is necessary to be able to design a split seal. Thus as part of this program, a small scale prototype wavy split seal was to be investigated. Thus, a design had to be made that would allow the carbon to be split (two places, 180° apart) and then clamped and still be able to transmit the wave-causing moments across the split. The first iteration for this design was to make a preload ring that would be pressed over the OD of the carbon ring. This preload ring would have to be able to apply a compressive load of 250 lb/in. radially at the centroid in order to offset the effect of the loads generated outward by the waviness inducer. Figure 3-19 shows the configuration of the band designed. The band has a slight taper ground into it as does the carbon, which allows for ease of press.

A three-dimensional SAPIV finite element analysis was then made on the carbon cross section without the band. Results showed a stiffness, GJ_{θ} , of 3527 lb-in². The design program used calculated a GJ_{θ} of 3183

TABLE 3-9
Weekly Average Leak Rate*
Test 129

Week	167 to 335	1679 hrs 1847 hrs	Predicted
Leakage	0.025 cc/min	0.032	0.4

*Based on total leakage for the week.



lb-in.², which is within 10 percent. It was therefore felt that the design program was satisfactory in its method of solution.

The next step was to re-run the three-dimensional SAPIV program with the inclusion of the preload ring. The resultant GJ_θ was 4368 lb-in.², a 24 percent increase. The band (Figure 3-11) was then made and pressed on a new carbon seal. The measured ninth harmonic waviness of the carbon without the preload ring was 50 μ in. With the preload ring on the carbon, the ninth harmonic waviness was measured at 32 μ in. a 36 percent reduction. The SAPIV finite element program predicted a 20 percent decrease in waviness with the addition of the band. From these results it was determined that the design method was sufficiently valid to serve as a useful tool.

From the preceding results, a new preload ring would need to be designed which would not restrict so much of the wave. Again, the three-dimensional SAPIV program was used and the design shown in Figure 3-20 was the result. This particular ring would only give a 12 percent increase in the stiffness, which would result in a reduction of ninth harmonic waviness from 50 μ in. to only 44 μ in.

The split seal itself, as shown in Figure 3-21, is to be machined out of two existing carbon seals. Internal moments are to be carried across the split sections by means of eight stainless steel dowel pins. These dowel pins were sized based on an analysis of the internal moment generated due to the induced waviness.

Referring back to Equation (3-32) the internal moment is given by

$$M_\theta = \frac{GJ_\theta}{R} (\phi' + \bar{v}') - \frac{EI^*}{R^3} (\phi''' + \bar{v}''') . \quad (3-52)$$

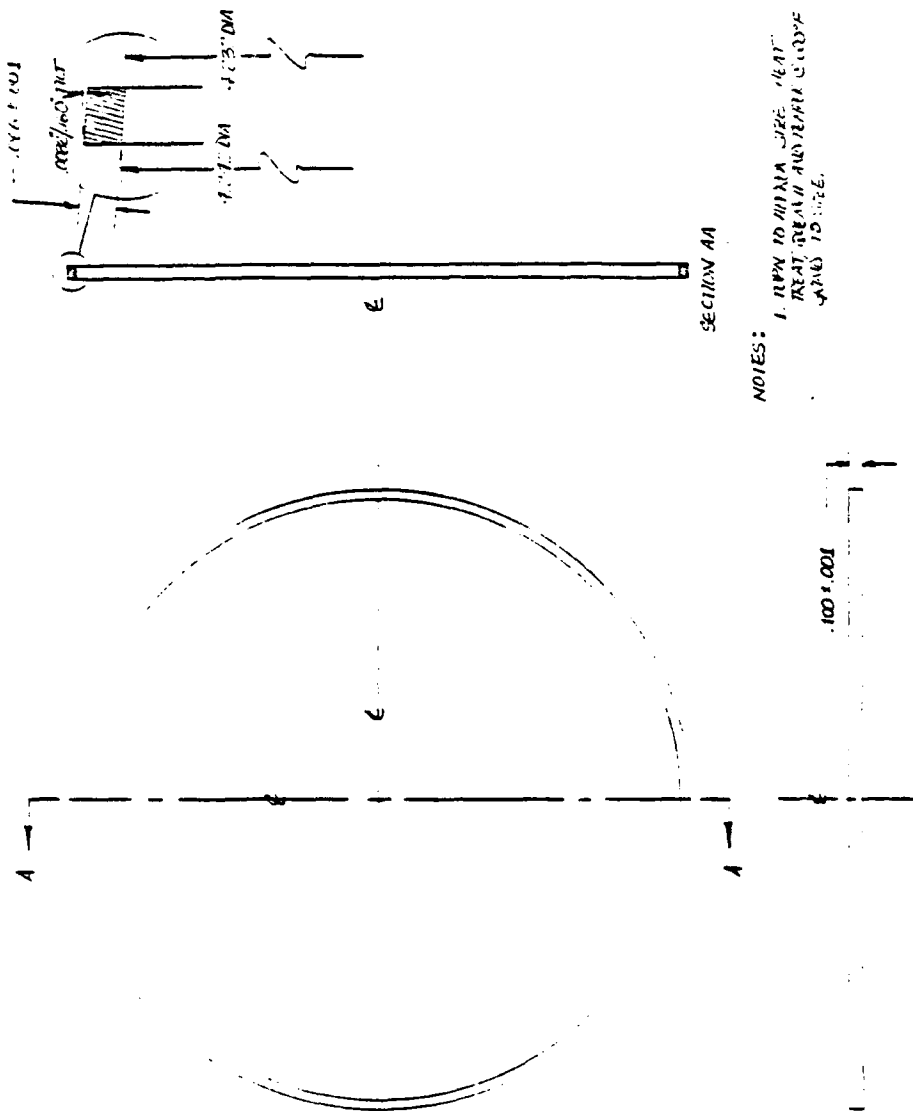


Figure 3-20. Modified Pre-load Ring.

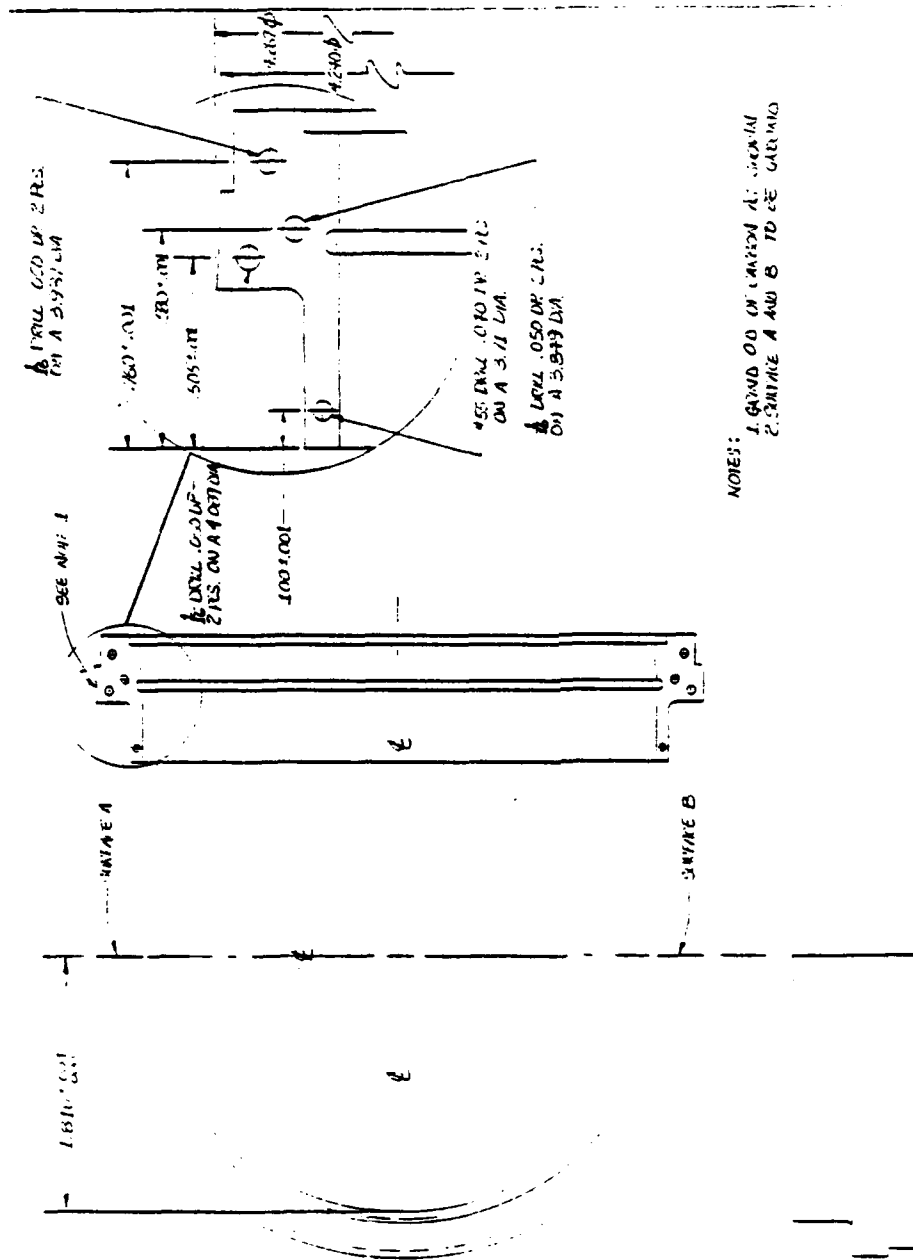


Figure 3-21. Split Ring Seal.

Considering warping gives,

$$\bar{v} = - \frac{m_{\theta} R^2}{EJ_x} C_1 \cos(n\theta) , \quad (3-53)$$

$$\bar{v}' = n \frac{m_{\theta} R^2}{EJ_x} C_1 \sin(n\theta) , \quad (3-54)$$

$$\bar{v}''' = -n^3 \frac{m_{\theta} R^2}{EJ_x} C_1 \sin(n\theta) , \quad (3-55)$$

and

$$\phi = \frac{m_{\theta} R^2}{EJ_x} C_2 \cos(n\theta) , \quad (3-56)$$

$$\phi' = -n \frac{m_{\theta} R^2}{EJ_x} C_2 \sin(n\theta) , \quad (3-57)$$

$$\phi''' = n^3 \frac{m_{\theta} R^2}{EJ_x} C_2 \sin(n\theta) , \quad (3-58)$$

where

$$C_1 = \frac{\frac{1}{A} + \frac{n^2}{B} + 1}{\frac{(n^2 - 1)^2}{A} + \frac{n^2}{B} (n^2 - 1)^2} \quad (3-59)$$

$$C_2 = \frac{\frac{n^2}{A} + \frac{n^4}{B} + n^4}{\left(\frac{n^2}{A} + \frac{n^4}{B} + n^4\right) \left(\frac{n^2}{A} + \frac{n^4}{B} + 1\right) - \left(\frac{n^2}{A} + \frac{n^4}{B} + n^2\right)^2} \quad (3-60)$$

Substitution of (3-54), (3-55), (3-57), and (3-58) into (3-52) gives

$$M_\theta = \frac{GJ_\theta}{R} \frac{nm_\theta R^2}{EJ_x} \left[[-C_2 \sin(n\theta) + C_1 \sin(n\theta) \right. \\ \left. - \frac{E\Gamma^*}{R^3} \frac{n^3 m_\theta R^2}{EJ_x} [C_2 \sin(n\theta) - C_1 \sin(n\theta)] \right] \quad (3-53)$$

Using the previous design data for the solid ring, the internal moment is

$$M_\theta = -7.65 \text{ in.-lb} \quad (3-54)$$

The dowels were sized to handle this moment.

Conclusions on Second Design

In many ways the performance of the second design was similar to that of the first design. Torque and leakage values were similar. Wear was significantly lower amounting to 355 μin . Since some taper did wear into this seal, it is expected that much of this wear took place early in the test, so the long term wear rate would be even better than on this test.

Theory predicts somewhat different values for friction torque than found experimentally but the hydrodynamic effect is nonetheless displayed. Leakage values are significantly lower than predictions. This is an advantage to seal operation but shows a significant weakness in the model.

An improvement is needed in the drive arrangement to get a positive engagement or improve the bond of the adhesive. Some improvements in oil plumbing reliability need to be made as there was one seep in the system.

Except for the problems mentioned the seal itself performed very well. At the wear rate measured, the wearing faces could be designed to meet a 150,000 hour life objective. The concept from the standpoint of wear reduction works very well. The question now becomes: would the waviness force applicator be sufficiently reliable? Certainly no significant problem was encountered in the 2000 hour test. The piston arrangement appears to be reliable. However, the O-rings and their associated plumbing introduces an element of chance failure or shortened life due to wear out which cannot be quantified. It can only be stated that the design would be more reliable absent these elements.

Thus, while the test and the design were both very successful, there is still a need to find a simple more reliable means to impose a wave.

CHAPTER 4

NINE WAVE SEAL--THIRD DESIGN

Early in the wavy seal investigation it became clear that a simple means of imposing a moving wave was essential to the ultimate success of the idea. While it was proven early in the test program that a moving wave would provide low friction and low wear along with low leakage, the greatest difficulties in designing, fabricating, and operating the wavy seal have been with the waviness causing device itself.

While a remedy to this problem has been sought all along, during the present contract period an extra effort was made to find a much simpler waviness device. To this end, the ideas explained in Chapter 8 on squeeze seals and bearings were conceived and analyzed. One of the conclusions as explained in Chapter 8 was that the squeeze seal (when the wave moves at shaft speed) is equivalent to the current wavy seal and therefore offers no advantage of further reduced wear. However, the squeeze seal does offer the possibility of facilitating the application of the wave. In fact, it happens that the squeeze seal as originally conceived is equivalent to forming the wave fixed in the hard face and rubbing against a flat soft face. This becomes then the basis for the third nine wave design.

The design is shown in Figures 4-1 and 4-2. The wave as shown in Figure 4-1 is ground into the hard face. The wave is similar to that used in present wavy seal designs just described in the previous chapters. Nine waves are again used so that the carbon does not

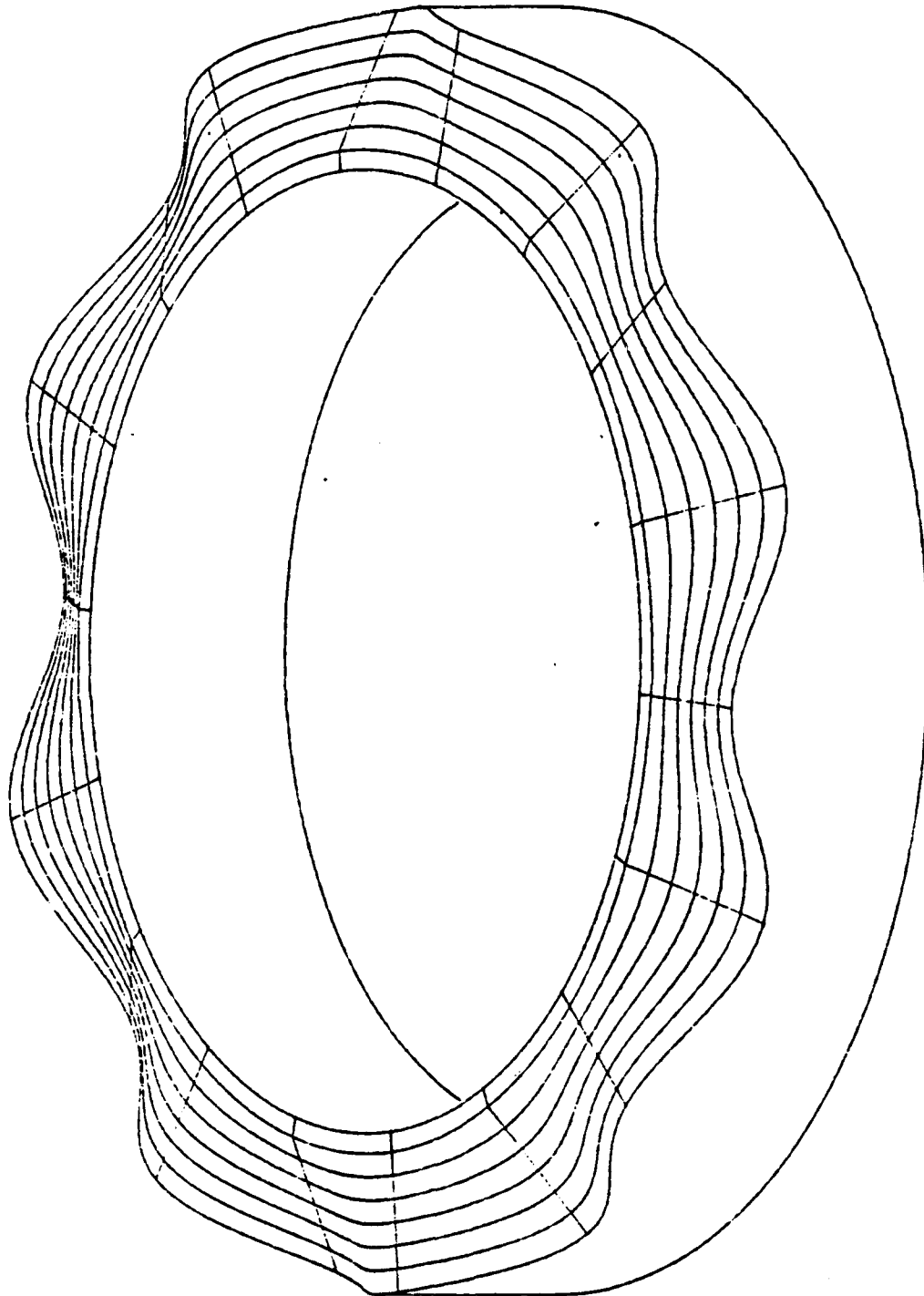


Figure 4-1. Wavy-Tilt-Dam Seal Face.
(Patent Pending)

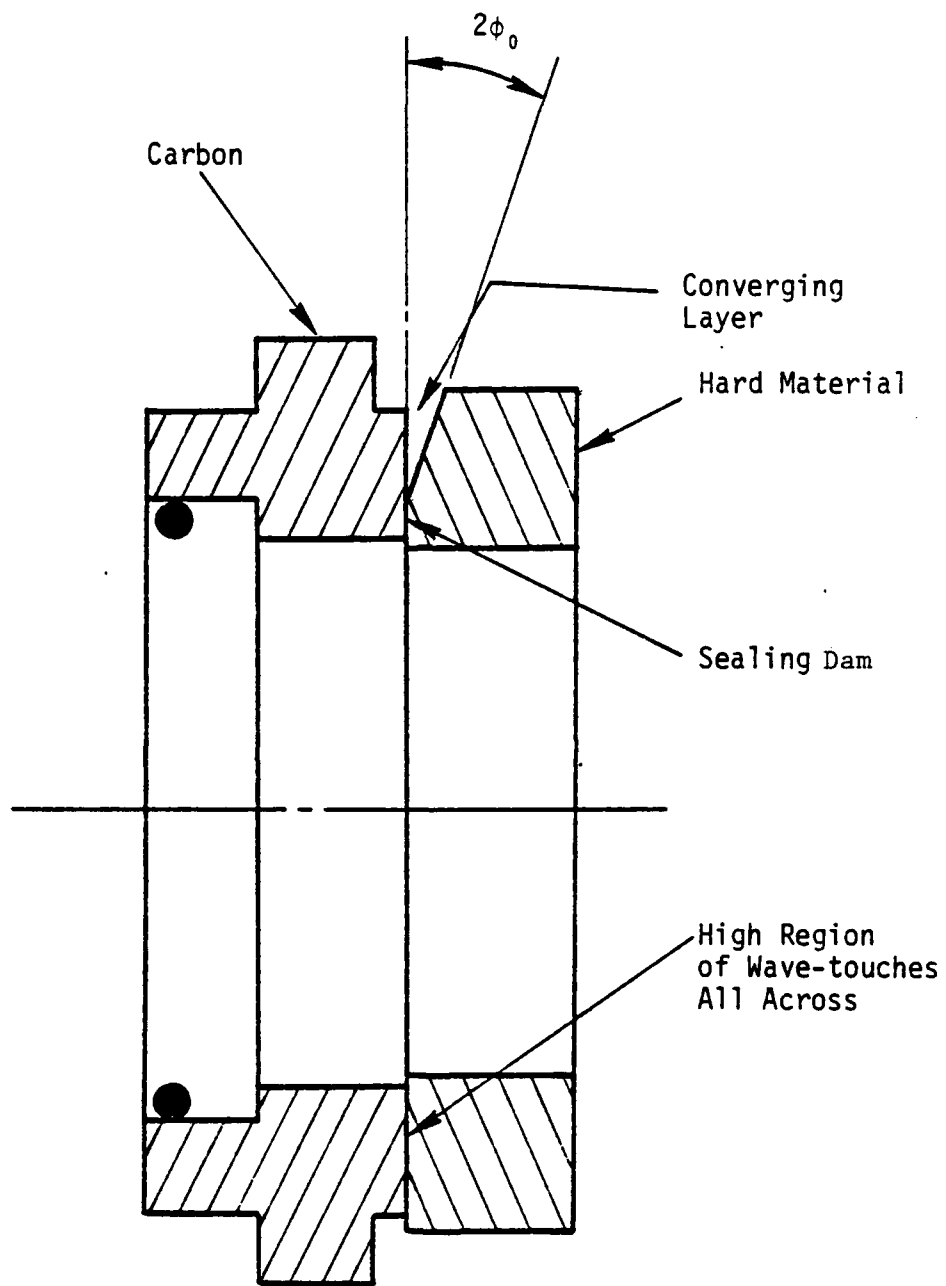


Figure 4-2. Wavy-tilt-Dam Seal.

significantly flatten out and thus eliminate the desired wave. The wavy-tilt design is used in order to obtain hydrostatic lift from the radial taper and hydrodynamic lift from the wavy part. The wavy tilt stops at some radius beyond which the seal face remains flat. This flat portion is the sealing dam and serves to minimize leakage flow.

In this design the carbon face is everywhere wiped by the high spots of the wave on the hard face. Thus the carbon must wear uniformly. The high spots on the wave do wear unevenly; that is, unlike design 2, wear is not spread out all across the hard face. However, the localized wear effect is on the hard face, not the carbon. The hard face is not wiped all over. Operation of the seal is identical to design 2.

Advantages

Compared to the moving wave, this seal design, once the parts are made, is as simple as current seals. No waviness drive is needed. Reliability should be very high. Because of the fact that with the present wavy design wear on the hard face is insignificant, it is expected that the same will hold true for this design. The carbon will wear slowly (same as at present). The hard face will wear very slowly such that the wave is preserved for the entire life of the seal. Long life and low leakage are expected just like for the current wavy seal.

Limitations

There is some difficulty in grinding the shape shown in Figure 4-1 into the hard face. While it is easy to deform carbon elastically to

get a wavy surface, this technique will not work for the high modulus hard face material.

The fact that the hard face is not wiped everywhere may allow debris or corrosion to build up in the low spots of the hard face. Also in some applications wear of the hard face presents an unknown factor. There is some evidence that hard face materials such as silicon carbide when lightly loaded (as in the case here) will experience only a few microinches of wear in a thousand hours. For applications where this is true for the fixed wave seal, then wear of the wave will present no problem. Experiments will answer the wear question as well as the corrosion buildup question.

One other limitation of this design is that special features must be incorporated so that radial misalignment can be accommodated. Figure 4-3 shows that with the normal seal design, a gap could result due to radial misalignment. The solution to this problem requires that the carbon and the hard face both be made wider as shown. The complications of solving the problem this way or using other approaches have not been evaluated as yet.

Wavy Seals

Wavy seals have been experimented with and proposed for use previously by other investigators [36,37]. The obvious question to be raised is why the wavy seal did not become a routine practice and what is different about the present proposal. First, the type of wave used and proposed here is a wavy tilt. The wavy tilt offers a sealing dam as well as hydrostatic support and hydrodynamic support. This type of

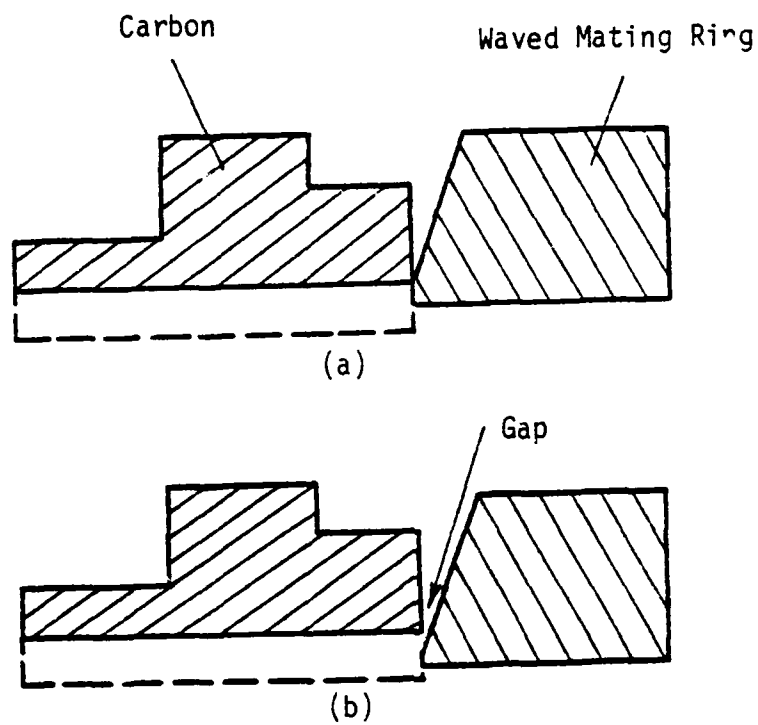


Figure 4-3. Effect of Contact Radius and Offset on Leakage.
(a) Properly Aligned Seals
(b) Result of Radial Offset

wave, as far as is known, has never been used before. Simply lapping a radially parallel wave into a seal may cause excessive leakage if the seal is stiff or nothing may happen if the seal is compliant [22]. Secondly, no consideration has been given previously to the relationship between stiffness and waviness. Early work in this project showed [4,5] that waves can be easily flattened out. It also showed that seal rings also tilt as they deform. Thus, imposing a wave on a seal ring without careful consideration of the deflection and net waviness may easily lead to undesirable contact geometries or unknown geometries. Deflection has been carefully considered in these designs. Thus in summary, the present work represents the first series of work where the wave shape has been carefully controlled, both in its creation and in operation, so that the desired effects will likely occur. Thus, the idea being proposed here, while not new as a general concept, is novel when considered carefully in detail, and it is these details which make the difference between successful operation and poor performance.

Seal Design

The third design is the result of only three modifications to the second design as shown in Figure 3-13. The first is the removal of the waviness inducer and spacer. This was done since the waviness is not imposed on the carbon for this design. The second modification was made on the method of driving the carbon. As pointed out before, the carbon rotated relative to the drive ring adapter in test no. 130, as a result of too much tangential load, due to friction torque of the seal, for the strength of the adhesive used. To eliminate this problem, four

#2-56 socket head cap screws were recessed into the carbon 0.05 inches to provide the additional drive capability along with the silicone adhesive.

The third modification was that of putting the desired wave on the hard mating face of the seal assembly. The technique for doing so is explained later, the performance for such a design follows.

Expected Performance

Seal performance is expected to be like that described in Chapters 2 and 3 for designs 1 and 2. However, some additional consideration was given to static (low speed) performance. Table 4-1 shows the static performance of the seal at various wave tilts. ϕ_0 is the tilt amplitude on the mating hard face and ϕ_{net} is the resulting tilt due to the conformability of the carbon. As can be seen, not a lot of friction reduction is gained by raising ϕ_m above 500 $\mu\text{in./in.}$ The leakage does, on the other hand, increase quite rapidly. This tradeoff was the basis for the selection of $\phi_0 = 500 \mu\text{m/in.}$ for the new design. Previous designs used a slightly lower waviness.

Waviness Grinding Apparatus

The procedure for putting the wave on the mating face is by means of grinding. Since the desired wave is of a special case, i.e., tilted in the radial direction, waved in circumferential direction, and having a circumferential strip of constant height at some given radial location, a unique method of grinding had to be devised.

TABLE 4-1

Seal Performance (Static)

$n = 9$ Speed = 1 RPM
 $P_o = 500$ psi

ϕ_o ($\mu\text{in/in}$)	ϕ_{net} ($\mu\text{in/in}$)	Q (cm^3/min)	T_q (in-lb)	μ	% fluid pressure load support
50	20	.09	120.6	.049	50
80	43	.10	113.9	.046	53
100	63	.11	106.0	.043	56
500	475	.33	68.6	.028	72
1000	967	.62	62.4	.025	74
1500	1461	1.07	59.7	.024	76
2000	1960	1.66	57.5	.023	77

Figures 4-4 and 4-5 show the grinding apparatus. The seal ring is mounted in a fixture and very precisely rotated about x-x in a lathe. The cam shown also rotates with the ring. The grinding wheel and its driver and the cam follower are all mounted to a plate which pivots about the axis y-y. The plate is flexure mounted as shown later so is actually very stiff relative to the base. However forces in the z direction produced by the cam follower and spring cause the plate to rotate (on its flexure mounts) about axis y-y. Therefore as the seal ring and cam turn about axis x-x, the cam follower causes the grinding wheel and plate assembly to oscillate about axis y-y. With the grinding wheel running about axis z-z, this action causes the wave of Figure 4-1 to be ground into the seal ring. The cam used has nine waves to produce the nine wave seal ring. Axis y-y is located at the sealing dam radius so that the waviness at that radius is zero.

Flexure Design

The flexure support system was chosen over bearings because of its high stiffness, zero looseness, and simplicity. The plate rotates only $\pm 500 \mu\text{m/m}$. Figure 4-6 shows the approximate pattern of flexure beams used to mount the plate. Point 0 is the pivot point whose location is assured by the four surrounding flexures. Flexures 5 and 6 provide additional stiffness to the mount.

Referring to Figure 4-6 summing moments about "0" give:

$$\sum M_0 = F_0 R_0 = F_1 R_1 + F_2 R_1 + F_3 R_1 + F_4 R_1 + F_5 R_2 + F_6 R_2 \quad (4-1)$$

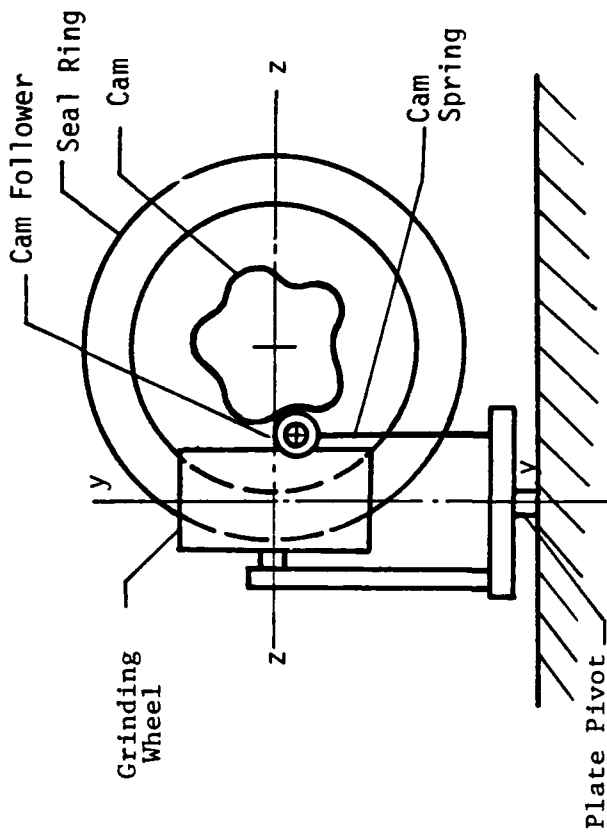
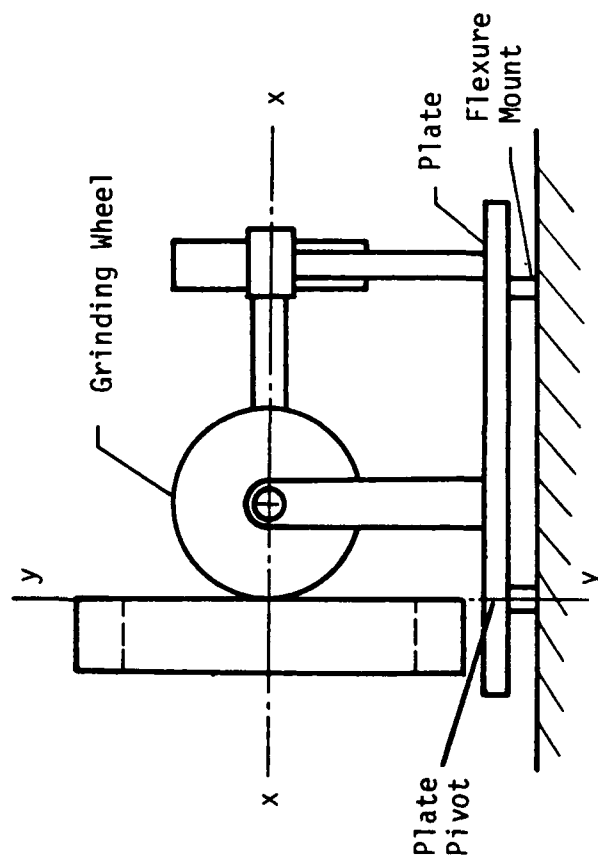
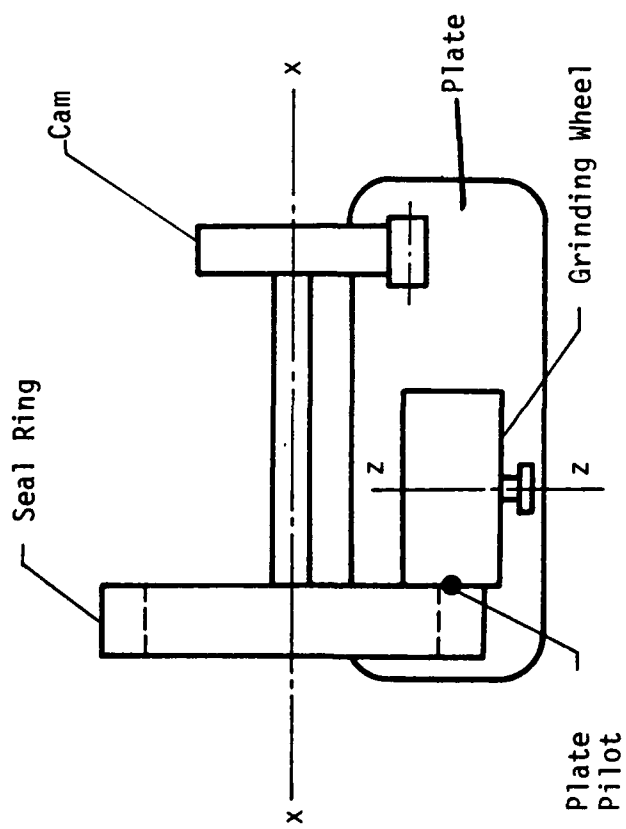


Figure 4-4. Grinding Apparatus Schematic.

(Patent Pending)

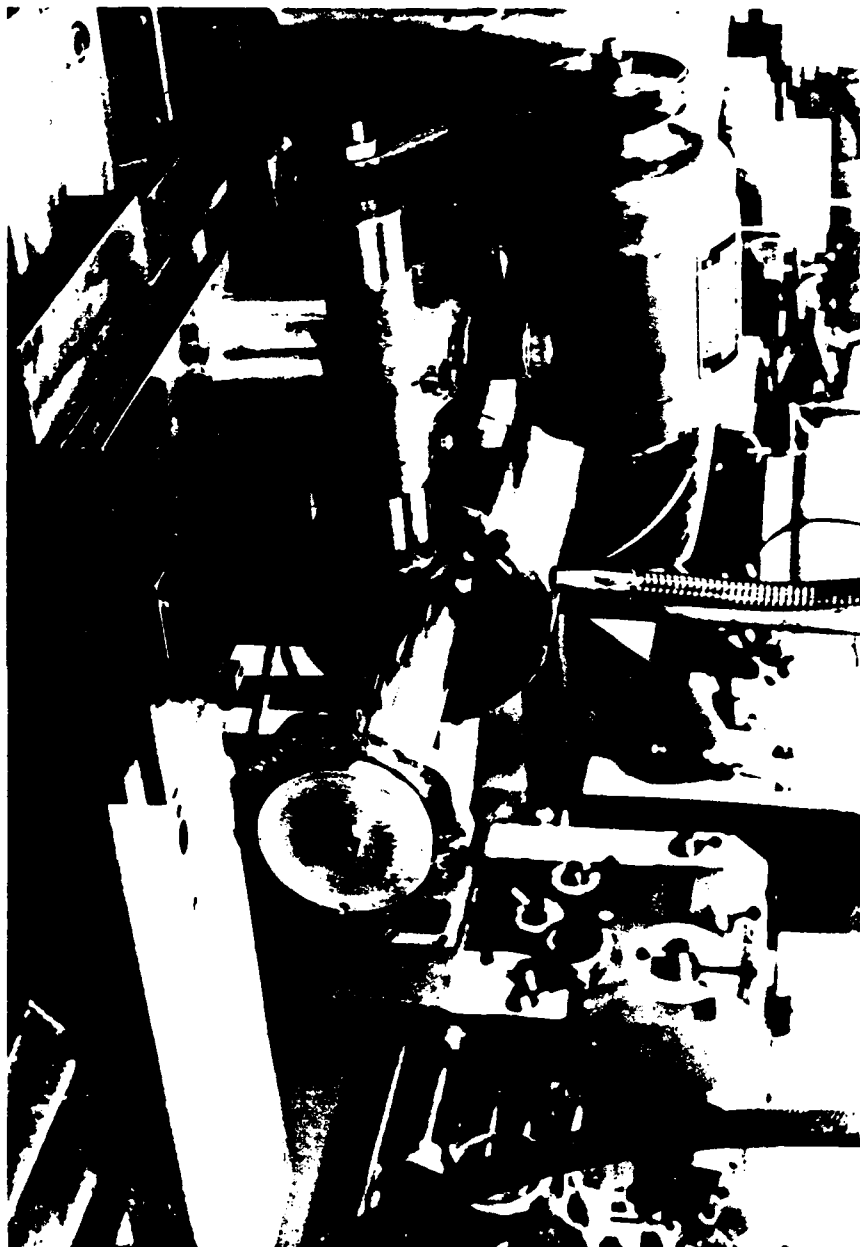


Figure 4-5. Grinding Apparatus.
(Patent Pending)

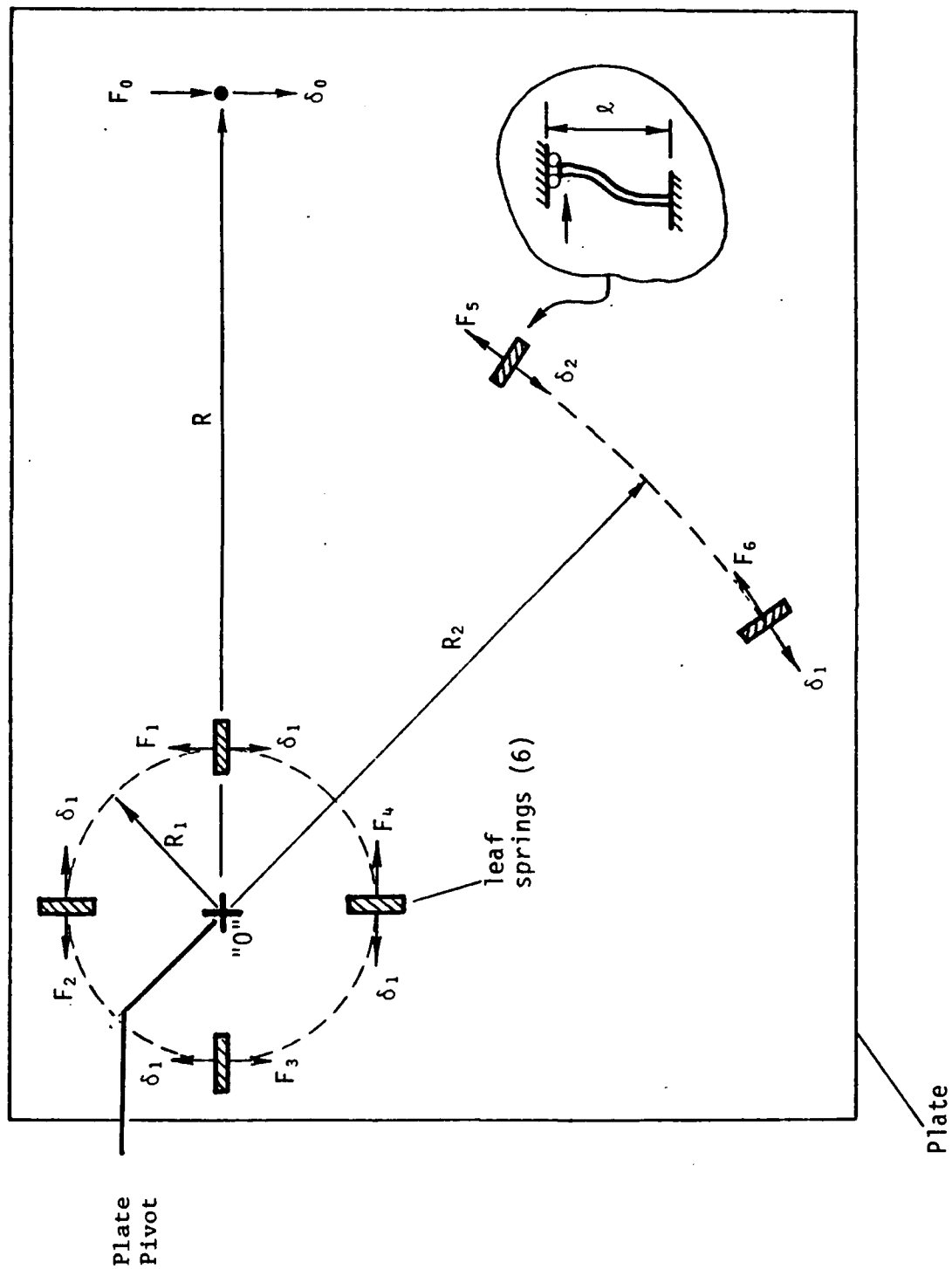


Figure 4-6. Flexure Mounts
(Patent Pending)

where

$$F = k\delta \quad (4-2)$$

and

$$\delta = R\theta \quad (4-3)$$

which gives

$$F = kR\theta . \quad (4-4)$$

Substituting (4-4) into (4-1) and assuming that all leaf springs have the same stiffness gives,

$$F_0 R_0 = 4kR_1^2 \theta + 2kR_2^2 \theta . \quad (4-5)$$

Now,

$$F_0 R_0 = k_T R_0^2 \theta \quad (4-6)$$

where k_T is the total effective spring constant of the system. Substitution of (4-6) into (4-5) gives

$$k_T = k \left[\frac{4R_1^2}{R_0^2} + \frac{2R_2^2}{R_0^2} \right] \quad (4-7)$$

The spring constant, k , for a beam fixed on one end and guided at the other end is [38],

$$k = \frac{12 EI}{l^3} . \quad (4-8)$$

The cam was designed to cause a 0.150 inch amplitude cam follower displacement. The stiffness was designed using the above relationships

so that the corresponding force produced by the cam follower spring produced the $\pm 500 \mu\text{m/m}$ rotation needed.

Waviness Profile Results

To test the grinding apparatus, a seal ring was machined out of 1018 steel to the dimensions of an existing seal ring. A 2 inch diameter, 60 grit, silicon carbide wheel was used for the grinding. The steel ring was then ground and a polar plot of waviness at the seal carbon outside radius (r_o) was taken. These results are shown in Figure 4-7. A Fourier analyses showed a ninth harmonic waviness of only 41 $\mu\text{in.}$ It was desired that a 75-100 $\mu\text{in.}$ (based on 500 $\mu\text{in./in.}$) wave be produced. The discrepancy was found to be due to the effect of stiffness of the connection between the upper grinding mount plate and the grinder itself. Some flexing was occurring at this connection and as a result the deflection of the grinding wheel was less than expected.

To correct for this, a calculation was made to re-size the input spring of the cam follower mechanism, making it stiffer. The steel ring was again ground and the resultant waviness plot is shown in Figure 4-8. The ninth harmonic waviness increased to 141 $\mu\text{in.}$ From these results it was concluded that the waviness grinding apparatus was indeed operating as designed and the next step was to grind a tungsten carbide for actual testing.

Grinding a tungsten carbide ring was done using a 2 inch diameter, 320 grit, diamond impregnated wheel. The results of this grinding is shown in Figure 4-9. The amplitude of the ninth harmonic wave is 134

POLAR PLOT OF WAVINESS

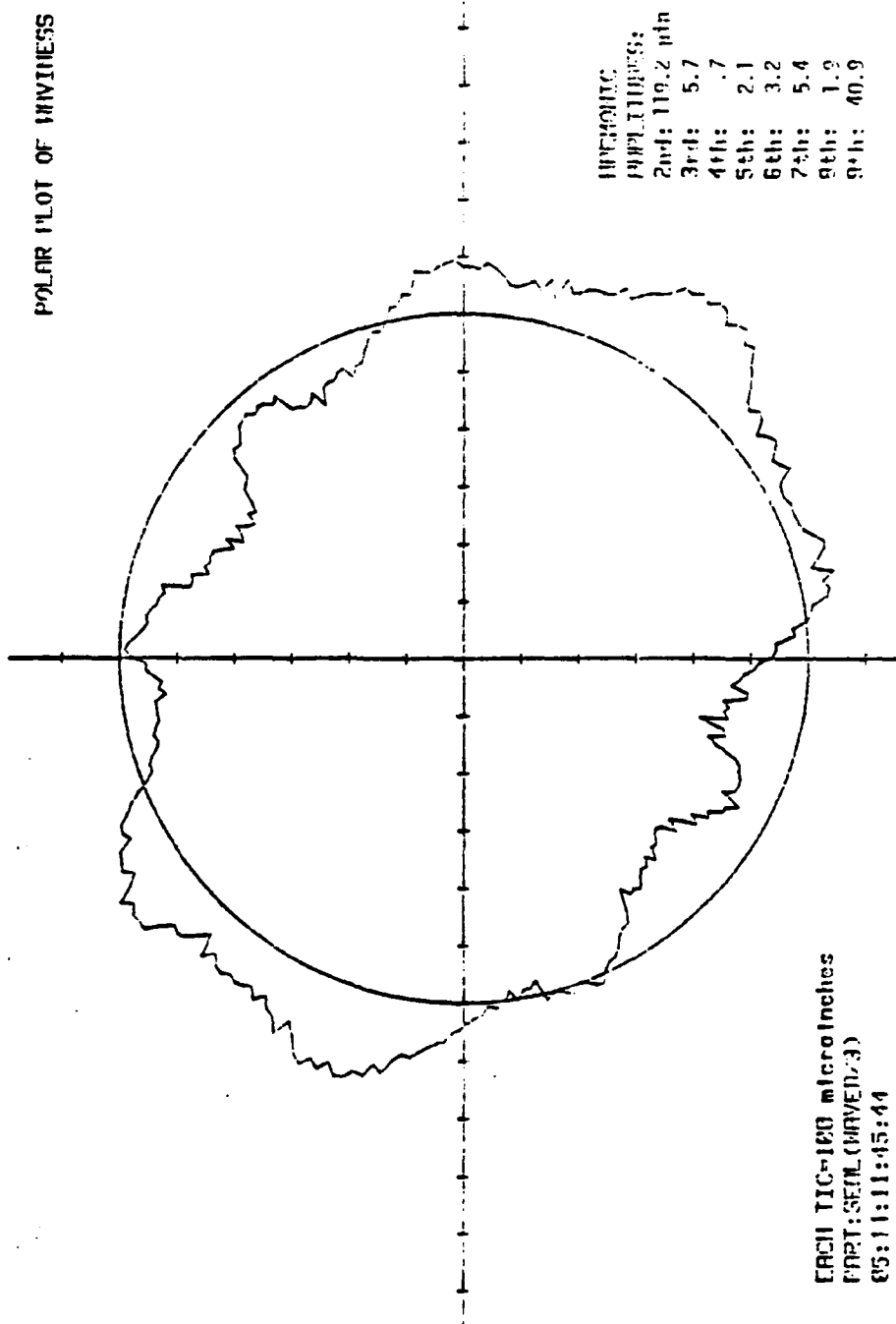


Figure 4-7. Waviness in Steel Ring--First Run.

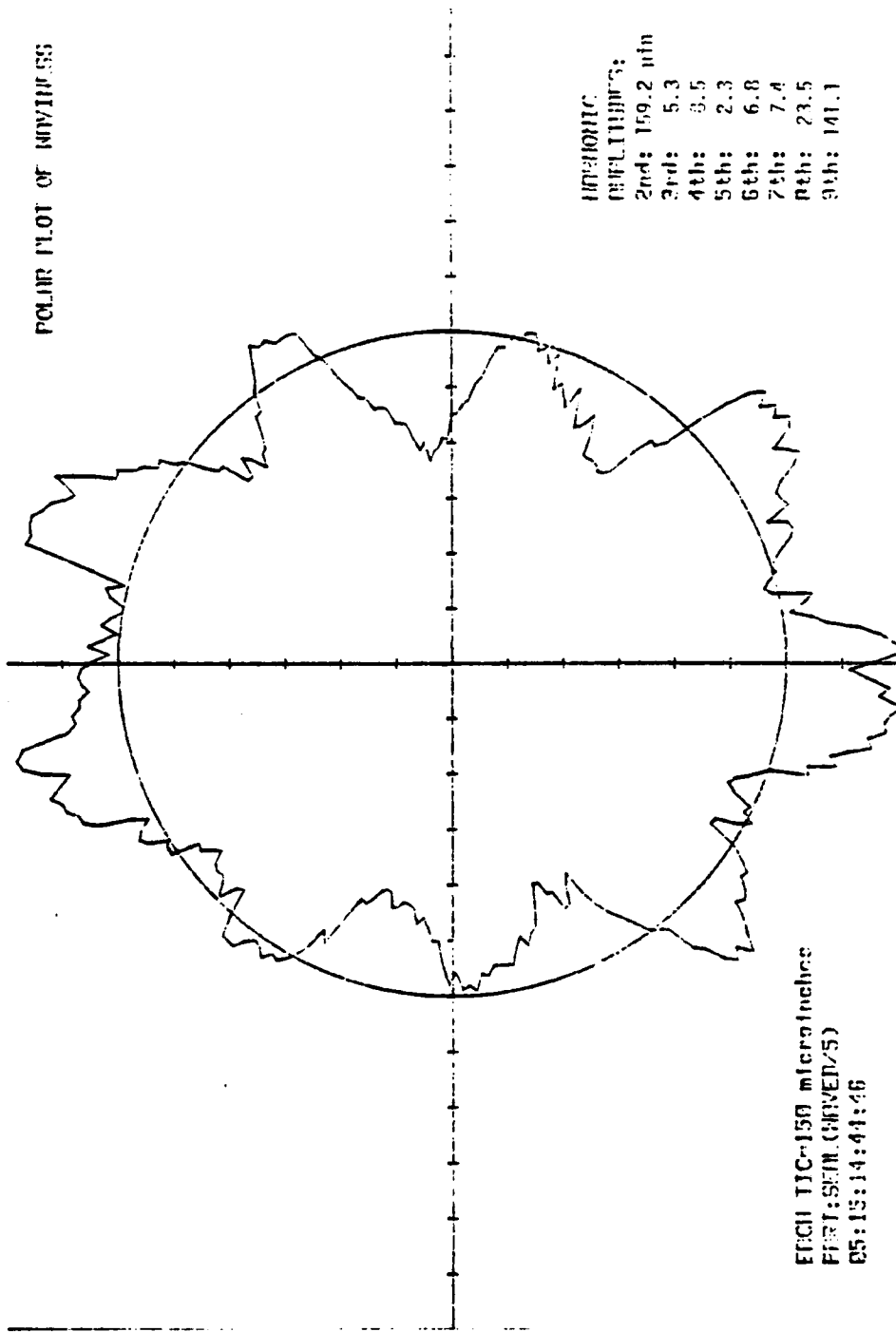


Figure 4-8. Waviness in Steel Ring--Second Run.

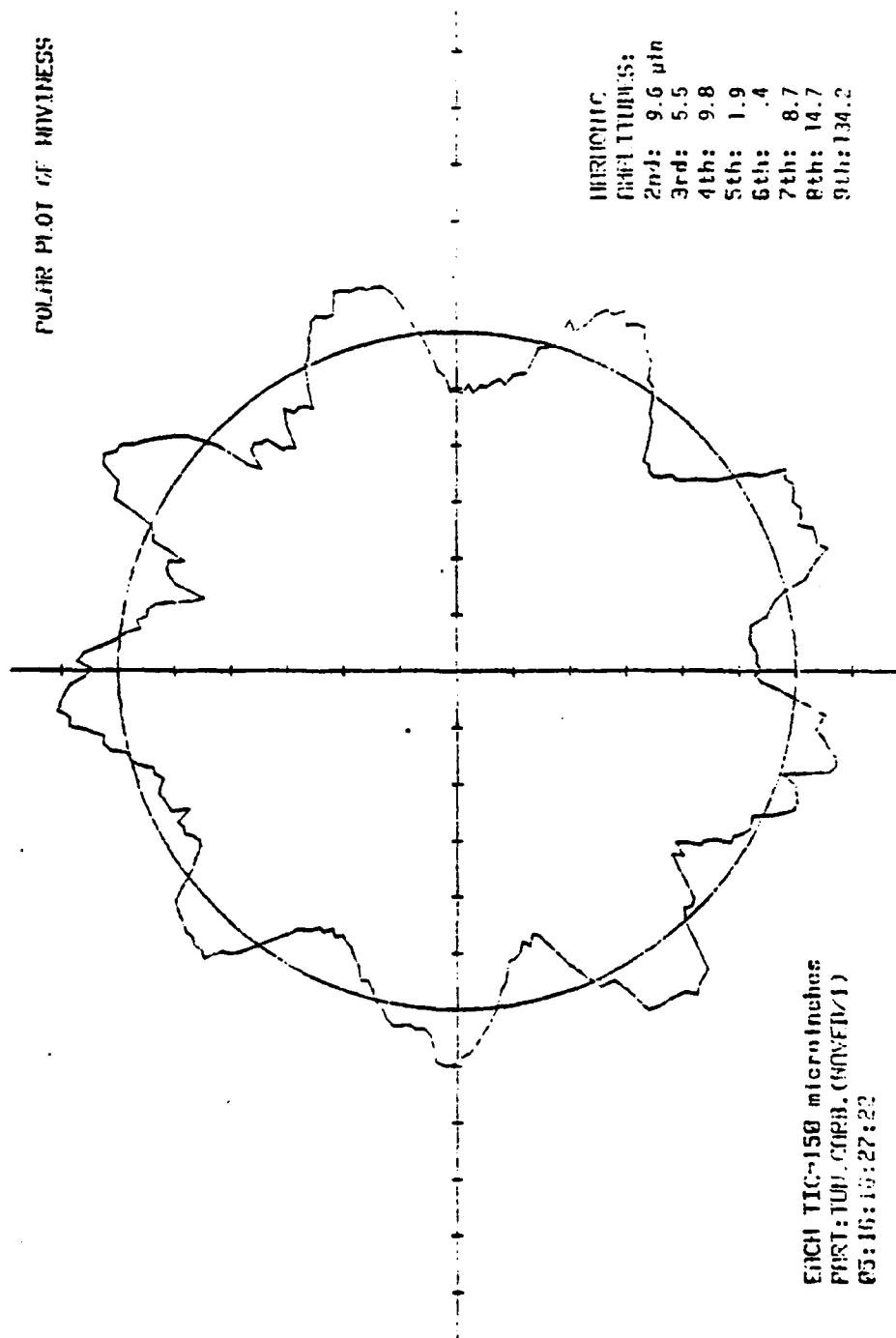


Figure 4-9. Waviness in w-c Ring.

μin., very close to that measured on the steel ring, which is good indication of repeatability. This tungsten carbide was used for the first 100 hour test.

100 Hour Test Results

Test No. 130 was the first 100 hour test run using the new concept of a wavy mating seal ring. The tungsten carbide ring had a wave 134 μin. amplitude at r_o . Figure 4-10 shows the performance during the test. The gap between approximately 62 hours and 112 hours was caused by a shutdown. The torque was, for the most part, unmeasurable, indicating nearly zero throughout the test. The leakage had some erratic behavior at the start, due to surface tension effects within the leakage collection passage. The average leakage for the approximate 100 hours of operation was about $3 \text{ cm}^3/\text{min}$.

Wear measurements were taken at four locations around the face of the carbon and compared to those taken before the test. Table 4-2 shows the results. The average wear was about 263 μin. It is believed that the wear occurred during the first few minutes of zeroing and test operation because the faces did not directly conform to each other at the start. Once the seal was in full operation, hydrodynamic effects lifts the faces apart, as evidenced by the low torque readings, and very little if any touching occurred.

Traces of the tungsten carbide face before and after the test showed no signs of wear.

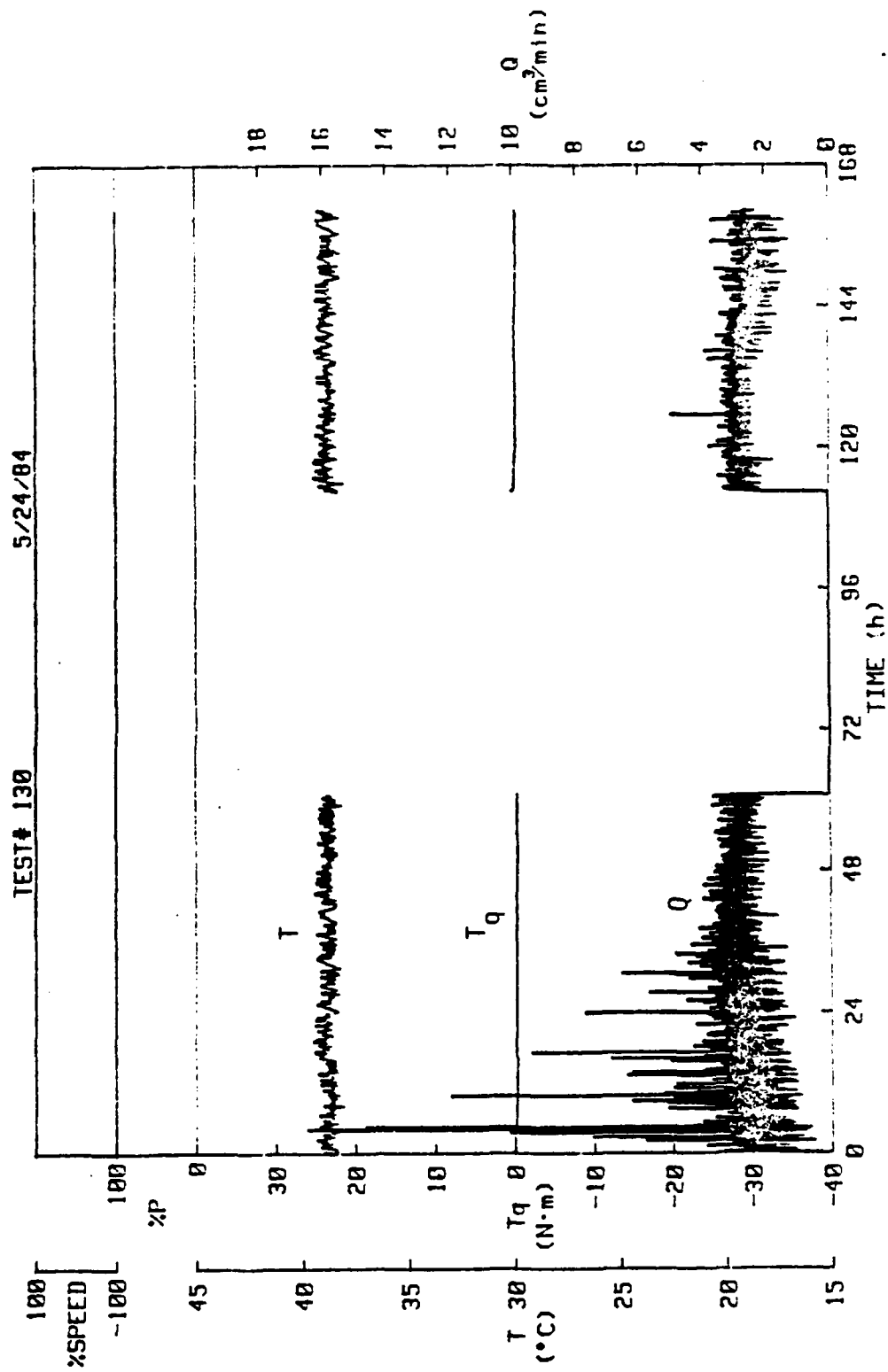


Figure 4-10. Test #130, 500 psi, 1800 RPM, Nine-wave Tungsten Carbide.

TABLE 4-2
Carbon Wear Measurements
Test # 130

Location	Wear
#1	200
#2	400
#3	250
#4	200

Average =263 μ in

Conclusions on Third Design

The first test using a fixed wave showed promising results. The wave was too large and needs to be reduced to lower the leakage rate. The surface roughness on the w-c was too large and there is concern that plowing wear will occur when the waviness is reduced. At the time of this writing short term tests are being made using other seal materials and waviness levels in anticipation of making a long term 2000 test. Thus, while results so far look very promising, too little experimental information is available to make final conclusions at this time.

CHAPTER 5

SEAL RING DEFLECTION

In the early days of seal design, seal ring deflection was essentially ignored. Seal rings were lapped flat and presumed to stay that way during operation. Now it is recognized that seal rings may undergo many different types of deflections, some causing excessive leakage and others allowing the seal to operate more effectively.

In this chapter no comprehensive review of seal deflection will be made; the subject would occupy several chapters of a book. Instead the more recent developments obtained under this research program will be presented in depth. These developments focus primarily on the prediction of whether or not seal rings will flatten out against each other in operation so as to minimize leakage. Theory is developed, tools are established, and many calculations on real seals are made and presented. It is thought that given these methods, much insight into the problem of seal leakage as related to deflection can be obtained, and such leakage can be minimized by well founded design changes.

Ring Finite Element

It was determined earlier in this investigation that a ring finite element was needed in which in-plane and out-of-plane forces and deflections are coupled by a non-zero product of inertia term. Such a finite element would serve as a basis for solving all types of ring deflection problems including a special class of two ring contact

problems. A survey of the literature indicated that the development of an element of this type had not been published.

Thus, such a ring finite element was derived and published in a previous report [6], thesis [28], and paper [19]. The ring FEM will be briefly described here so that subsequent developments in this report are more clear. Figure 5-1 shows a ring with all of the possible types of loading shown. Figure 5-2 shows a segment of the ring and all of the internal and external moments and forces acting on it. Figure 5-2 serves as a basis for writing six equations of equilibrium for the ring segment. Using these equations and stress resultant-displacement relationships [34,39] allows one to derive the generalized relationship between the stress resultants and the displacements for an arbitrary segment of the ring--the stiffness matrix in the finite element method. Figure 5-3 shows the element, the end forces and moments, and the displacements. The generalized displacement and force vectors are given by

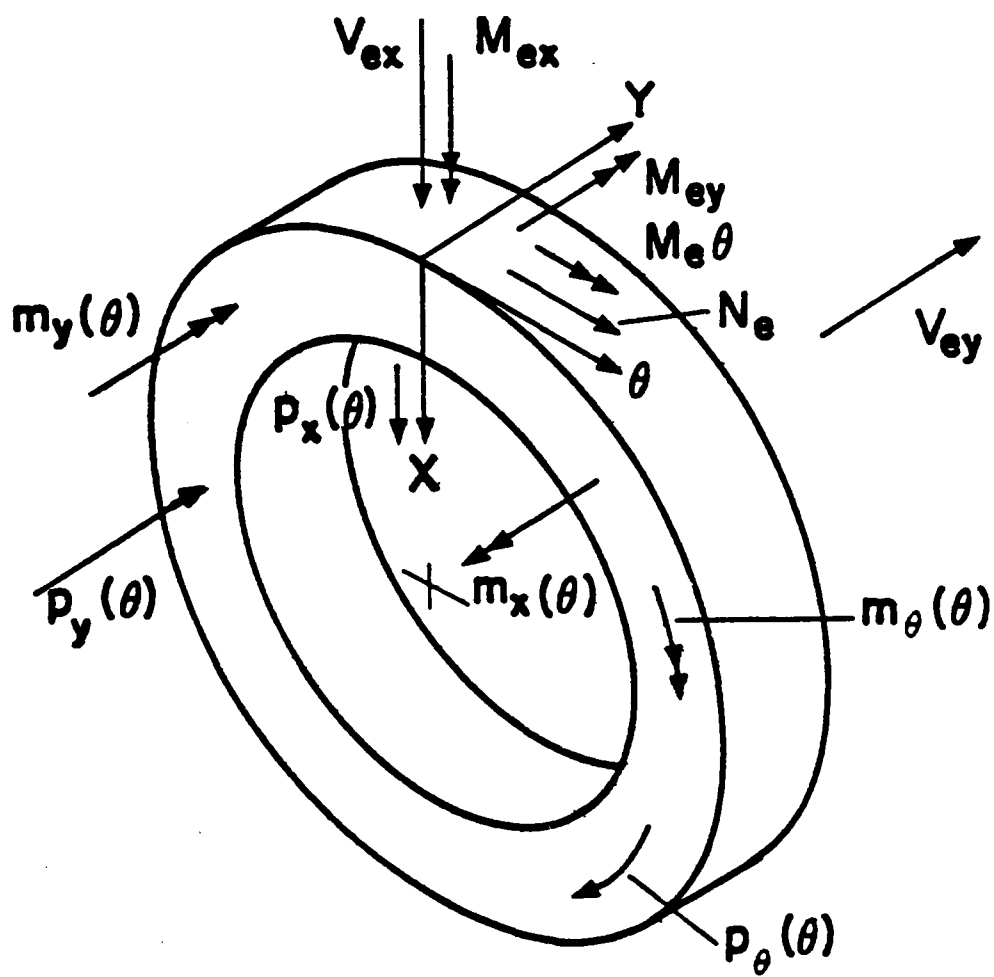


Figure 5-1. Loads on a Ring.

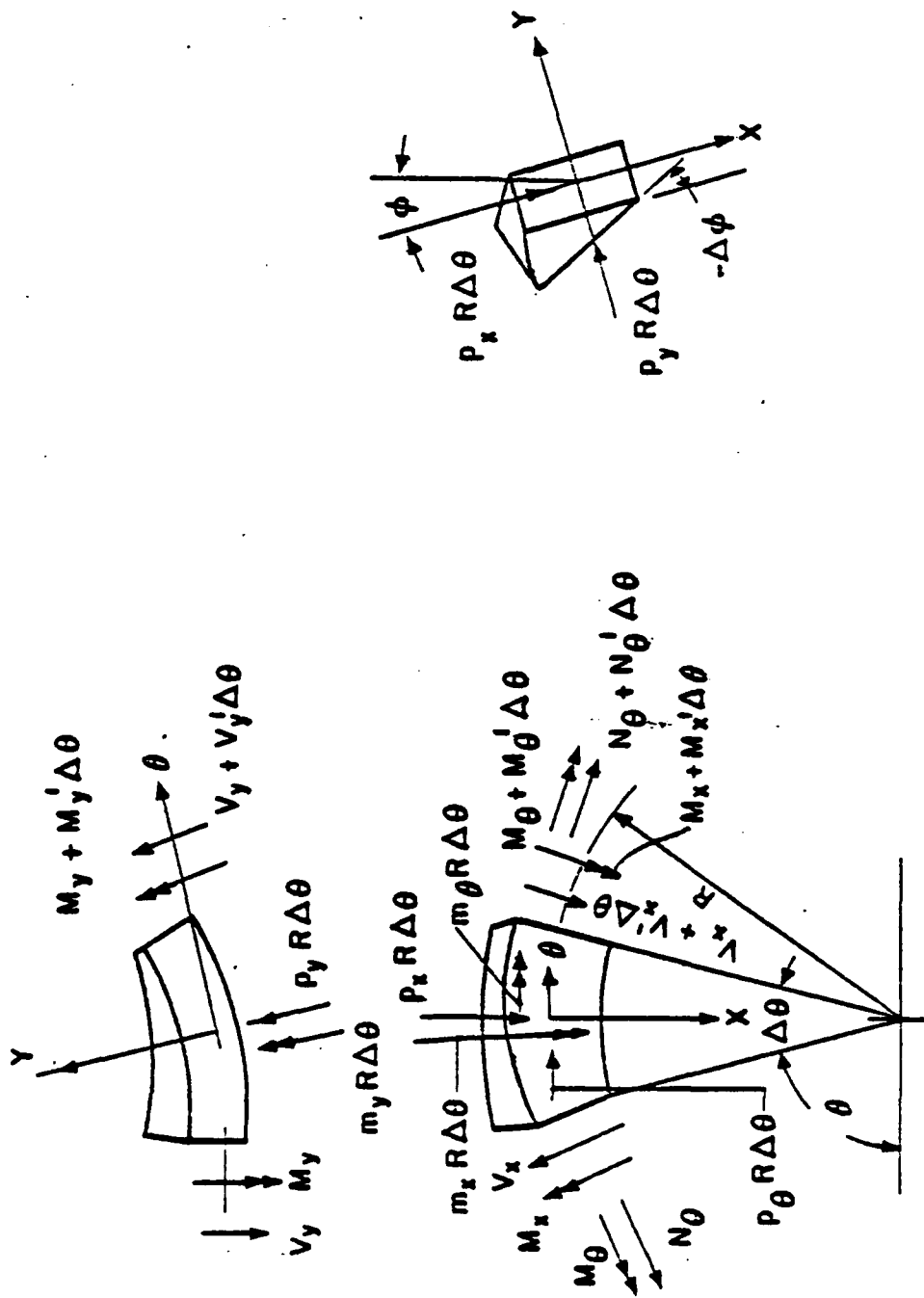


Figure 5-2. Ring Geometry and Definition.

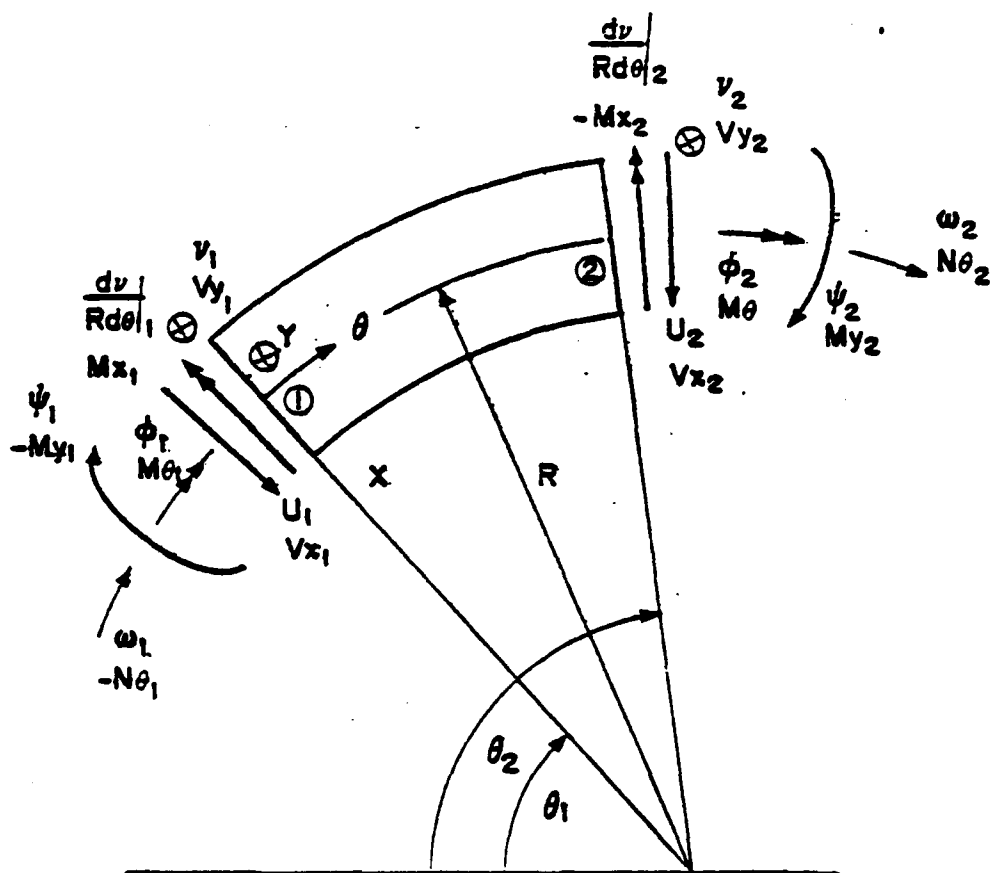


Figure 5-3. Ring Finite Element.

$$[\delta] = \begin{bmatrix} u_1 \\ v_1 \\ w_1 \\ , \\ \frac{v_1}{R} \\ \psi_1 \\ \phi_1 \\ u_2 \\ v_2 \\ w_2 \\ , \\ \frac{v_2}{R} \\ \psi_2 \\ \phi_2 \end{bmatrix} \quad (5-1)$$

$$[F] = \begin{bmatrix} -V_{x_1} \\ -V_{y_1} \\ -N_{\theta_1} \\ M_{x_1} \\ -M_{y_1} \\ -M_{\theta_1} \\ V_{x_2} \\ V_{y_2} \\ N_{\theta_2} \\ -M_{x_2} \\ M_{y_2} \\ M_{\theta_2} \end{bmatrix} \quad (5-2)$$

The stiffness matrix $[K]$ is defined by the relationship between forces and displacements:

$$[F] = [K] [\delta] . \quad (5-5)$$

The previous work provided two essential matrix equations shown on the next two pages. Equations [5-4] and [5-5] relate the force and deflection vectors to section properties and the arbitrary constants of the solution to the differential equations. Repeating

$$[A] [G] = [\delta] , \quad (5-3)$$

$$\frac{EJ}{R^2} [D] [G] = [F] . \quad (5-4)$$

$$\begin{aligned}
 & \left[\begin{array}{cccccccccccc}
 0 & 0 & \frac{V}{R} \cos \theta_1 & -\frac{V}{R} \sin \theta_1 & \frac{2U}{R} \sin \theta_1 & \frac{2U}{R} \cos \theta_1 & 0 & \frac{V}{R} \cos \theta_1 & -\frac{V}{R} \sin \theta_1 & 0 & 0 & 0 \\
 0 & -\frac{1}{RA} & 0 & 0 & 0 & 0 & 0 & 0 & 0 & 0 & 0 & 0 \\
 0 & 0 & \frac{V}{R} \sin \theta_1 & \frac{V}{R} \cos \theta_1 & -\frac{2U}{R} \cos \theta_1 & \frac{2U}{R} \sin \theta_1 & 0 & \frac{V}{R} \sin \theta_1 & \frac{V}{R} \cos \theta_1 & 0 & 0 & 0 \\
 0 & 0 & -\frac{1}{A} \sin \theta_1 & -\frac{1}{A} \cos \theta_1 & 0 & 0 & 0 & -\frac{1}{A} \sin \theta_1 & -\frac{1}{A} \cos \theta_1 & 0 & 0 & 0 \\
 0 & 0 & -V \sin \theta_1 & -V \cos \theta_1 & 2U \cos \theta_1 & -2U \sin \theta_1 & -U & -V \sin \theta_1 & -V \cos \theta_1 & 0 & 0 & 0 \\
 0 & -\frac{1}{A} & -\frac{1}{A} \cos \theta_1 & \frac{1}{A} \sin \theta_1 & 0 & 0 & 0 & \frac{1}{A} \cos \theta_1 & \frac{1}{A} \sin \theta_1 & 0 & 0 & 0 \\
 0 & 0 & -\frac{V}{R} \cos \theta_2 & \frac{V}{R} \sin \theta_2 & -\frac{2U}{R} \cos \theta_2 & -\frac{2U}{R} \sin \theta_2 & 0 & -\frac{V}{R} \cos \theta_2 & \frac{V}{R} \sin \theta_2 & 0 & 0 & 0 \\
 0 & \frac{1}{RA} & 0 & 0 & 0 & 0 & 0 & 0 & 0 & 0 & 0 & 0 \\
 0 & 0 & -\frac{V}{R} \sin \theta_2 & -\frac{V}{R} \cos \theta_2 & \frac{2U}{R} \cos \theta_2 & -\frac{2U}{R} \sin \theta_2 & 0 & -\frac{V}{R} \sin \theta_2 & -\frac{V}{R} \cos \theta_2 & 0 & 0 & 0 \\
 0 & 0 & \frac{1}{A} \sin \theta_2 & \frac{1}{A} \cos \theta_2 & 0 & 0 & 0 & \frac{1}{A} \sin \theta_2 & \frac{1}{A} \cos \theta_2 & 0 & 0 & 0 \\
 0 & 0 & V \sin \theta_2 & V \cos \theta_2 & -2U \cos \theta_2 & 2U \sin \theta_2 & U & V \sin \theta_2 & V \cos \theta_2 & 0 & 0 & 0 \\
 0 & \frac{1}{A} & \frac{1}{A} \cos \theta_2 & -\frac{1}{A} \sin \theta_2 & 0 & 0 & 0 & \frac{1}{A} \cos \theta_2 & -\frac{1}{A} \sin \theta_2 & 0 & 0 & 0
 \end{array} \right] [D] = [G] = [r]
 \end{aligned}$$

$$\begin{aligned}
 & \begin{bmatrix} G_1 \\ G_2 \\ G_3 \\ G_4 \\ G_5 \\ G_6 \\ G_7 \\ G_8 \\ G_9 \\ G_{10} \\ G_{11} \\ G_{12} \end{bmatrix} = \begin{bmatrix} -V_{x1} \\ -V_{y1} \\ -M_{01} \\ M_{x1} \\ -M_{y1} \\ -M_{01} \\ V_{x2} \\ V_{y2} \\ N_{02} \\ -M_{x2} \\ M_{y2} \\ M_{02} \end{bmatrix}
 \end{aligned}$$

$\frac{EJ}{R^2} x$

$$U = \frac{J_x}{J_{xy}} - \frac{J_{xy}}{J_y} \cdot V = \frac{J_x}{J_{xy}} \left(1 + \frac{1}{A} \right) - \frac{J_{xy}}{J_y}$$

Equation (5-4)

Given (5-5) above Equations (5-4) and (5-5) may be solved to give

$$[K] = \frac{EJ}{R^2} [D] [A]^{-1} . \quad (5-6)$$

Then the stiffness matrix for the coupled ring problem is readily evaluated as the product of known matrix D and the inverse of known matrix A. It is useful to observe that if $\theta_2 - \theta_1$ is constant and the section properties are constant, [K] is constant--a very useful property when assembling a global stiffness matrix.

If $J_{xy} = 0$ the above equations may not be used directly to solve for K. However one may numerically allow $J_{xy} \rightarrow 0$ to get a satisfactory result. Also, when $J_{xy} = 0$ one actually gets two uncoupled cases, and these have been solved exactly and are presented in References [6], [19], and [28].

The element derived is readily assembled into a closed circular ring or a segment of a ring. Elements of various sizes ($\theta_2 - \theta_1$) can be used as needed. Assembly is straightforward in that no coordinate transformations are needed. Also, just as in common straight beam elements, the element derived is the exact solution to the governing equations. Thus, if no distributed loads are present, then the element size can be as large as possible while still accommodating concentrated loads. If distributed loads are present, then element size must be made relatively small to give a good approximation. On two examples for distributed load cases, it was found that 40 elements gives deflection results within a few percent of that predicted by exact theory.

Beyond these points, assembly is straightforward and needs no further discussion.

Comparison to Other Results

Solutions to two coupled problems were found in the literature and have been used to check the accuracy of the derived element.

The first check is based on earlier analytical work by Lebeck [34]. The problem is shown in Figure 5-4. A circular ring is subjected to two tangential concentrated loads. These are equilibrated by a distributed tangential load. For $J_{xy} \neq 0$, an out-of-plane deflection v is produced by this loading. For a completely arbitrary selection of section properties and using 36 elements, the deflection was found using the element derived. The deflection was compared to that found analytically and given in Reference [34]. Agreement was exact to four significant places. Since both the previous solution and the present solution are based on the same equations, this agreement verifies the correctness of the derivations of the element only, not the beam theory used.

A second check problem was taken from the work of Meck [40]. Using the energy method, Meck derives the equation for the three-dimensional displacement of the end of a curved nonsymmetrical beam (other end fixed) caused by completely arbitrary loading. The problem is shown in Figure 5-5. Using the coupled finite element given and arbitrary section properties, a solution was found for all displacements u , v , w , ψ , v' , ϕ for each of the loads shown. Agreement with Meck's solution

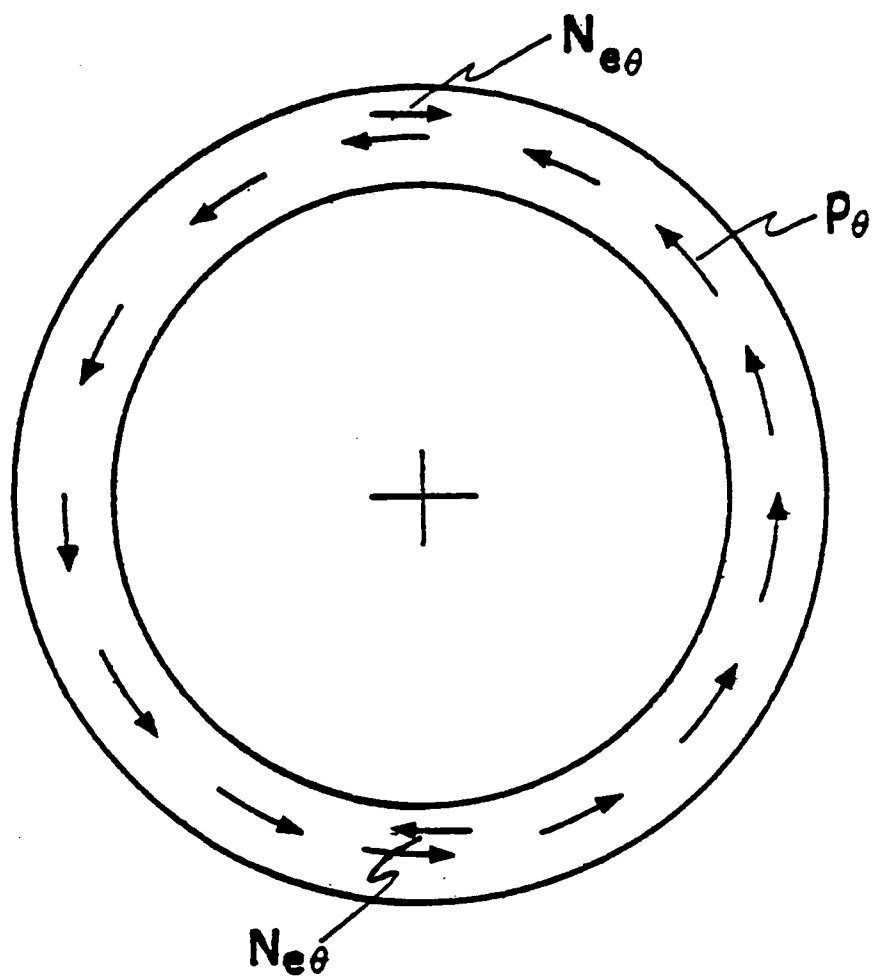


Figure 5-4. Check Problem 1.

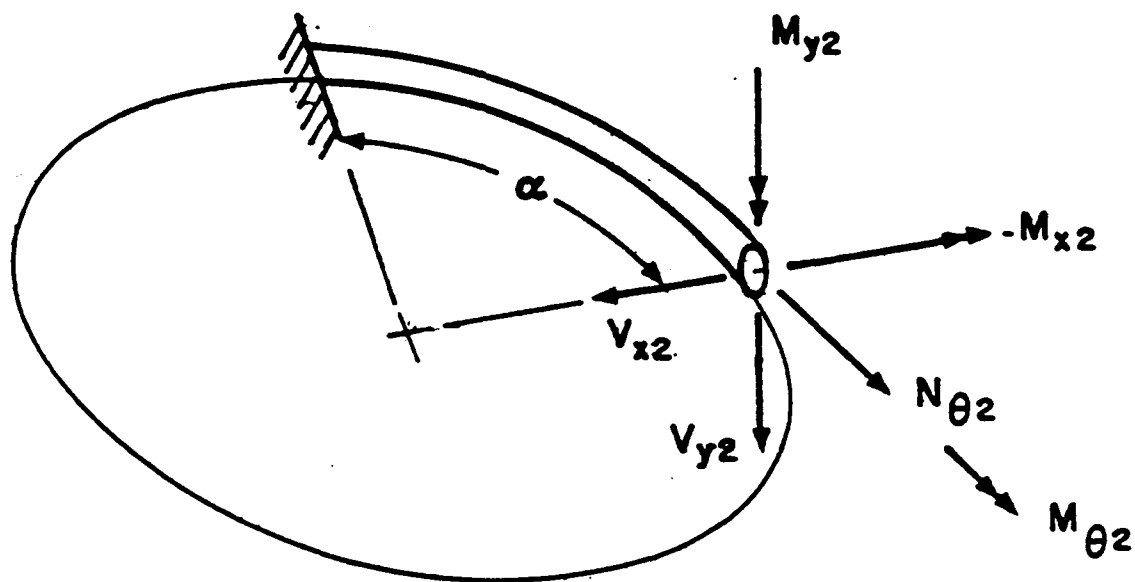


Figure 5-5. Check Problem 2.

was excellent thus verifying the correctness of the beam theory used relative to other beam theory as well as the finite element itself.

A third check was made experimentally using the aluminum ring shown in Figure 5-6. The ring was loaded by two radial loads through the centroid. Before loading and after loading measurements of deflection were made at locations (1), (2), and (3) perpendicular to the surface using a precision rotary table and precision displacement transducer. Data was analyzed to pick out on the second harmonic net distortions in each case. The experimental results are shown in Table 5-1. The results show clearly how an out-of-plane deflection is produced by the in-plane load.

The theoretical results shown were obtained using an 8 element assembly and finding the second harmonic component of the various displacements. Agreement on the in-plane deflections is reasonable. The out-of-plane deflection error is larger. There are numerous sources of error when comparing this experiment to theory. First, the shear center does not coincide with the centroid as is assumed in the theory. Second, exact material properties for the particular alloys used were not measured. Third, only an approximate formula was used to find J_{θ} . Fourth, the measurements themselves are good only to a few percent. Thus, without considerable refinement in theory and experiment, one probably has as good an agreement as can be expected; and the experiment does generally verify the coupling described by the theory.

The coupled ring finite element will now be used in various developments.

$$\begin{aligned}
 R &= 48.65 \text{ mm} \\
 a &= 83.97 \text{ mm}^2 \\
 J_x &= 1052 \text{ mm}^4 \\
 J_y &= 369.7 \text{ mm}^4 \\
 J_{xy} &= 149.5 \text{ mm}^4 \\
 J_\theta &= 869.9 \text{ mm}^4
 \end{aligned}$$

Aluminum Ring

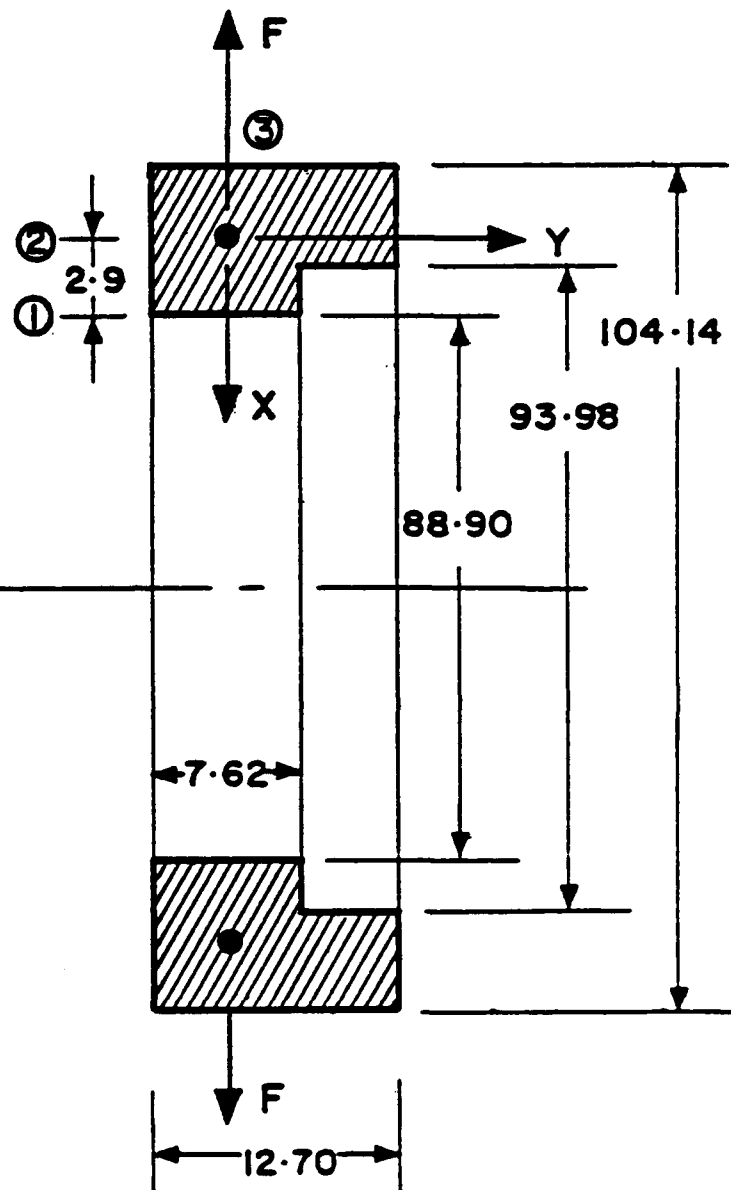


Figure 5-6. Test Ring.

TABLE 5-1
Experimental Results
111 N Radial Load - 2nd Harmonic Amplitudes

Location	1	2	3	2 - 1
Experiment	6.27 μm	6.63 μm	41.48 μm	0.36 μm
Theory	4.90	5.21	36.93	0.31
% Error	22	21	11	14

Single Ring Deflection by Formulas

While finite element methods are acknowledged as being accurate and capable of handling complex problems, many times it is useful to be able to calculate ring deflection for seals using easily available formulas. Many formulas have been derived during the early period of this seal research and are presented in Reference [34] and [39]. Since that time several additional useful formulas have been derived and the formulas have been checked using the FEM program described in the next section.

The complete set of formulas available at present is presented here. These cases are those which have been found to be useful for seal ring calculations. Particular note of equilibrium loads must be made. They were chosen to most closely represent what will happen as faces touch together in a seal. The deflection formulas for v only (out of plane of the face) deflections are given in Table 5-2. ϕ is also important but such formulas are not available at this time.

Single Ring Deflections by FEM

The ring element described in the first section of this chapter was used as a basis for writing a general ring finite element program. The program is described in detail in Appendix B. The program is set up for equal element size with constant section properties but can be readily modified to deal with variable elements so that a non-axisymmetric ring as well as non-axisymmetric loads can be considered. The program calculates the Fourier series coefficients of the primary deflections in recognition of the importance of these deflections for

TABLE 5-2

Ring Deflection

Equation for displacement

Equilibrium Load

Type Load

Waviness

None required

Two equal and opposite moments M_{ex} located 180° apart

$$\begin{bmatrix} v_2 \\ v_4 \\ v_6 \end{bmatrix} = \begin{bmatrix} 0.0707 \\ 0.0028 \\ 0.0005 \end{bmatrix} \frac{M_{ex} R^2}{EJ_x} (1 + \Lambda)$$

$$v = \frac{M_{ex} R^2}{EJ_x} (1 + \Lambda) \left[\sin \theta + \left(\frac{\pi}{2} - \theta \right) \cos \theta \right]$$

$$0 \leq \theta \leq \pi$$

None required

Two equal and opposite moments M_{ex} located 180° apart

$$\begin{bmatrix} v_2 \\ v_4 \\ v_6 \end{bmatrix} = \begin{bmatrix} 0.212 & 0.283 & 0.356 & 0.424 \\ 0.013 & 0.014 & 0.016 & 0.017 \\ 0.003 & 0.003 & 0.004 & 0.004 \end{bmatrix} \frac{M_{ex} R^2}{EJ_x}$$

$$v = \frac{M_{ex} R^2}{EJ_x} \left[\Lambda \left(\frac{\theta}{\pi} - \frac{1}{2} \right) + \frac{1}{2} \Lambda \cos \theta + \frac{1}{4} (\Lambda + 1) \left(\theta - \frac{\pi}{2} \right) \sin \theta \right]$$

$$0 \leq \theta \leq \pi$$

Sinusoidal varying pressure p_y around face

$$\begin{bmatrix} v_2 \\ v_4 \\ v_6 \end{bmatrix} = \begin{bmatrix} 0.0354 \\ 0.0050 \\ 0.0014 \\ 0.0006 \end{bmatrix} \frac{M_{ex} R^2}{EJ_x} (1 + \Lambda)$$

$$v = \frac{M_{ex} R^2}{EJ_x} (1 + \Lambda) \left[-\frac{1}{8\pi} \theta^2 \cos \theta + \frac{\pi}{16} - \frac{1}{4} \theta \sin \theta - \frac{\pi}{16} \cos \theta + \frac{1}{4\pi} \theta \sin \theta + \frac{1}{4} \theta \cos \theta \right]$$

$$0 \leq \theta \leq 2\pi$$

Sinusoidal varying pressure p_y around face

$$\begin{bmatrix} v_2 \\ v_4 \\ v_6 \end{bmatrix} = \begin{bmatrix} 0.106 & 0.141 & 0.177 & 0.212 \\ 0.018 & 0.022 & 0.025 & 0.028 \\ 0.006 & 0.007 & 0.008 & 0.008 \\ 0.003 & 0.003 & 0.003 & 0.004 \end{bmatrix} \frac{M_{ex} R^2}{EJ_x}$$

$$v = \frac{M_{ex} R^2}{EJ_x} \left[-\frac{1}{8} + \frac{\Lambda}{8} \theta^2 \sin \theta - \frac{\Lambda}{2} + \frac{\Lambda}{2\pi} \theta + \frac{\Lambda}{2} \cos \theta + \frac{(1 + \Lambda)}{4} \theta \sin \theta - \frac{\Lambda}{2\pi} \theta \cos \theta \right]$$

$$0 \leq \theta \leq 2\pi$$

Sinusoidal varying plus a constant pressure p_y at face

$$\begin{bmatrix} v_2 \\ v_4 \\ v_6 \end{bmatrix} = \begin{bmatrix} 0.0532 & 0.0707 & 0.0884 & 0.1060 \\ 0.0061 & 0.0072 & 0.0083 & 0.0094 \\ 0.0016 & 0.0018 & 0.0020 & 0.0021 \\ 0.0006 & 0.0006 & 0.0007 & 0.0007 \end{bmatrix} \frac{V_{ey} R^3}{EJ_x}$$

$$v = \frac{V_{ey} R^3}{EJ_x} \left[-\frac{(\Lambda + 1)}{8\pi} \theta^2 \cos \theta - \frac{\Lambda}{4\pi} \theta^2 + \frac{\Lambda}{2} \theta + \frac{3\Lambda + 1}{4\pi} \theta \sin \theta + \frac{\Lambda + 1}{4} \theta \cos \theta \right]$$

$$-\frac{(\Lambda - 1)\pi}{16} - \frac{(3\Lambda + 1)}{4} \sin \theta + \frac{(\Lambda - 1)\pi}{16} \cos \theta$$

$$0 \leq \theta \leq 2\pi$$

n Forces V_{ey} located $\alpha = 2\pi/n$ apart

$$\begin{bmatrix} v_2 \\ v_4 \\ v_6 \end{bmatrix} = \begin{bmatrix} 0.0531 & 0.0707 & 0.0884 & 0.1061 \\ 0.0061 & 0.0072 & 0.0083 & 0.0094 \\ 0.0016 & 0.0018 & 0.0019 & 0.0021 \\ 0.0006 & 0.0006 & 0.0007 & 0.0007 \end{bmatrix} \frac{V_{ey} R^3}{EJ_x}$$

$$v = \frac{V_{ey} R^3}{EJ_x} \left[\frac{\Lambda}{2} \left(\frac{\theta}{\alpha} - 1 \right) + \frac{\Lambda + 1}{4} \frac{(6\alpha \sin \theta + 2\alpha) - \pi \sin \theta}{2(\cos \theta - 1)} - (1 - \cos \theta) \right]$$

$$0 \leq \theta \leq \alpha$$

$$\frac{3\Lambda + 1}{4} \sin \theta - \frac{\Lambda + 1}{4} \frac{\sin \theta}{\cos \theta - 1} - \frac{\Lambda + 1}{4} \theta \sin \theta - \frac{\Lambda + 1}{4} \theta \cos \theta$$

$$0 \leq \theta \leq \alpha$$

TABLE 5-2 (Cont.)

Ring Deflection

Face pressure of n waves of amplitude p_0	None required	$A = 2 \quad 4 \quad 6 \quad 8$				$v = \frac{p_0 R^4}{E J_x} \left[\frac{A + n^2}{n^2(n^2 - 1)^2} \right] \cos n\theta$	$0 \leq \theta \leq 2\pi$
		v_2	v_3	v_4	v_6		
Two equal and opposite loads V_{ex} located 180° apart ²	None required	0.1667	0.2222	0.2778	0.3333	$v = \frac{V_{ex} R^3}{4E(J_{xy} - J_{xy}^2)} \left(-8 \cos \theta - \frac{\pi}{2} + \sin \theta + \frac{\pi}{2} \cos \theta \right)$	$0 \leq \theta \leq \frac{\pi}{2}$
		0.0191	0.0226	0.0260	0.0295		
Two tangential loads N_{ey} located 180° apart ²	Uniform tangential friction p_0 at face	0.0050	0.0056	0.0061	0.0067	$v = \frac{N_{ey} R^3}{E(J_{xy} - J_{xy}^2)} \left(-\frac{\theta}{\pi} + \frac{1}{2} - \frac{1}{4} \theta \sin \theta + \frac{\pi}{8} \sin \theta - \frac{1}{2} \cos \theta \right)$	$0 \leq \theta \leq \frac{\pi}{2}$
		0.0019	0.0020	0.0022	0.0023		
Two moments M_{ey} located 180° apart ²	Uniform tangential friction p_0 at face	0.0009	0.0009	0.0010	0.0010	$v = \frac{M_{ey} R^2}{E(J_{xy} - J_{xy}^2)} \left[\frac{\theta}{\pi} + \frac{1}{2} (\cos \theta - 1) \right]$	$0 \leq \theta \leq \pi$
		0.0707	0.0028	0.0005	0.0000		
Two equal and opposite loads V_{ex} located 180° apart ²	None required	0.0707	0.0028	0.0005	0.0000	$v = \frac{V_{ex} R^3}{E(J_{xy} - J_{xy}^2)}$	$0 \leq \theta \leq \frac{\pi}{2}$
		0.0354	0.0007	0.0000	0.0000		
Two tangential loads N_{ey} located 180° apart ²	Uniform tangential friction p_0 at face	0.0354	0.0007	0.0000	0.0000	$v = \frac{N_{ey} R^3}{E(J_{xy} - J_{xy}^2)}$	$0 \leq \theta \leq \frac{\pi}{2}$
		0.0106	0.011	0.003	0.003		

¹ $J_{xy} = 0$.

² $J_{xy} \neq 0$.

³ These are the fundamental harmonics only, e.g., v_4 is not the fourth harmonic resulting from two loads, but in the fourth harmonic resulting from 4 loads.

⁴ Only the fundamental harmonic exists here.

seal analysis. It is generally useful for all types of ring deflection problems and has served as a basis for the more complex contact problems discussed later. The program was used to check the Fourier series coefficients of the previous section.

Footnote 1 in the table denotes all cases where the assumption of $J_{xy} = 0$ has been used to simplify the derivation. These formulas are valid even if $J_{xy} \neq 0$ for a first approximation of deflection. The last three cases (footnote 2) were derived specifically for coupled problems where $J_{xy} \neq 0$ couples the in-plane deflection to the out-of-plane deflection or waviness.

In addition to the formulas for deflection the tables provide Fourier series coefficients for each case. The series terms are either all even or all odd depending on the loading case. These data show at a glance the magnitude of the wave caused by a disturbance. The first harmonic is not shown since it represents tilt of the ring which has no meaning in a seal since seals are self-aligning.

Formulas for $n - V_{ey}$ and n waves of p_{yo} are used to study the flattenability or conformability of seal rings. This subject is discussed in detail later in this report.

Simple Seal Contact Model

Seal performance is dramatically affected when the waviness/stiffness combination is such that the faces do not contact all around and leakage gaps develop. Prediction of such gaps in the general case is a complex problem as will be discussed in detail in the next section.

Very useful but simple models of seal contact can also be developed. On the basis of simple harmonic waviness, two formulas given in Table 5-2 can be used to estimate flattenability of a ring in face contact. There are at least two reasonable assumptions which might be made as to how the net flattening load is distributed. For seal rings where waviness is fairly large, the load might be considered as being concentrated at n points equally spaced as shown in Figure 5-7 for $n = 2$. In this case the individual loads are given by

$$V_{ey} = \frac{F_{net}}{n} . \quad (5-7)$$

Case b in Figure 5-7 shows the net flattening load distributed continuously and sinusoidally. It is readily shown that the maximum amplitude the sine wave can have while maintaining zero or greater contact pressure all around the seal is

$$P_{amplitude} = \frac{F_{net}}{2\pi R} . \quad (5-8)$$

Both of these distributions serve to make useful calculations. The discrete load case gives the maximum flattening which can occur. It is used to compute the net wave in a case where contact will remain at n points even after the wave has been flattened somewhat. On the other hand, case b represents a load which causes the maximum possible waviness for a given net load while still maintaining contact all around the seal. That is, case b can be used to calculate the maximum wave amplitude allowable before some separation of the rings occurs.

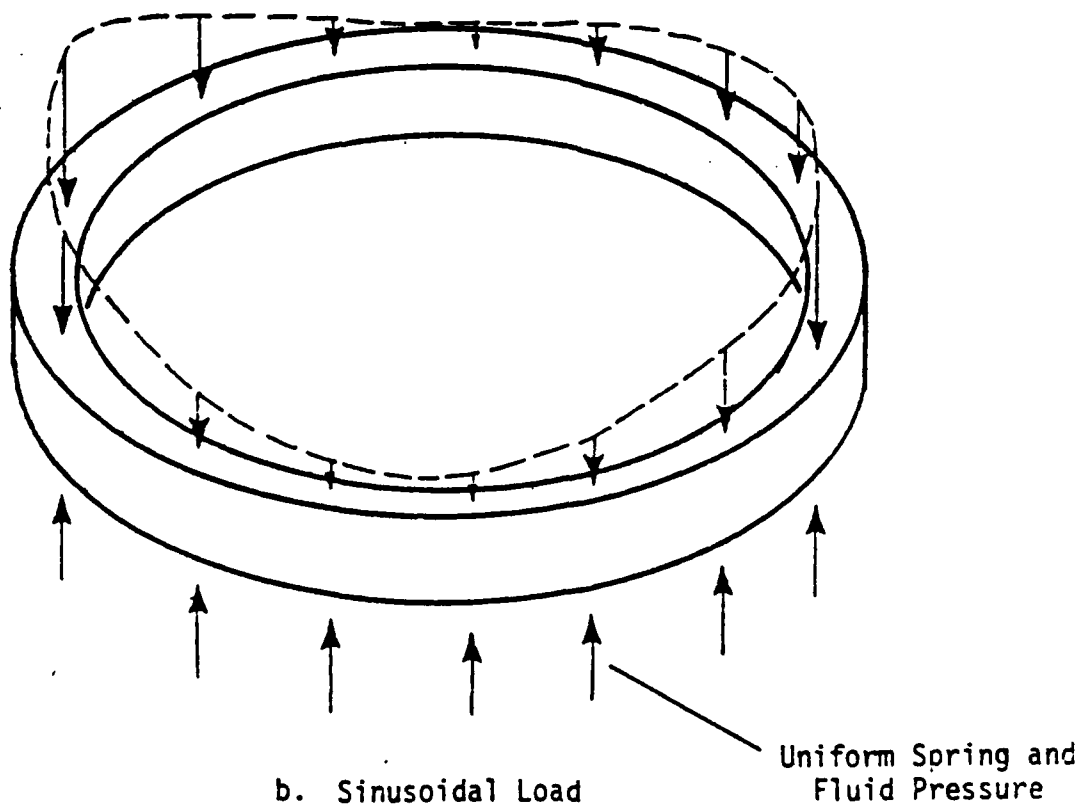
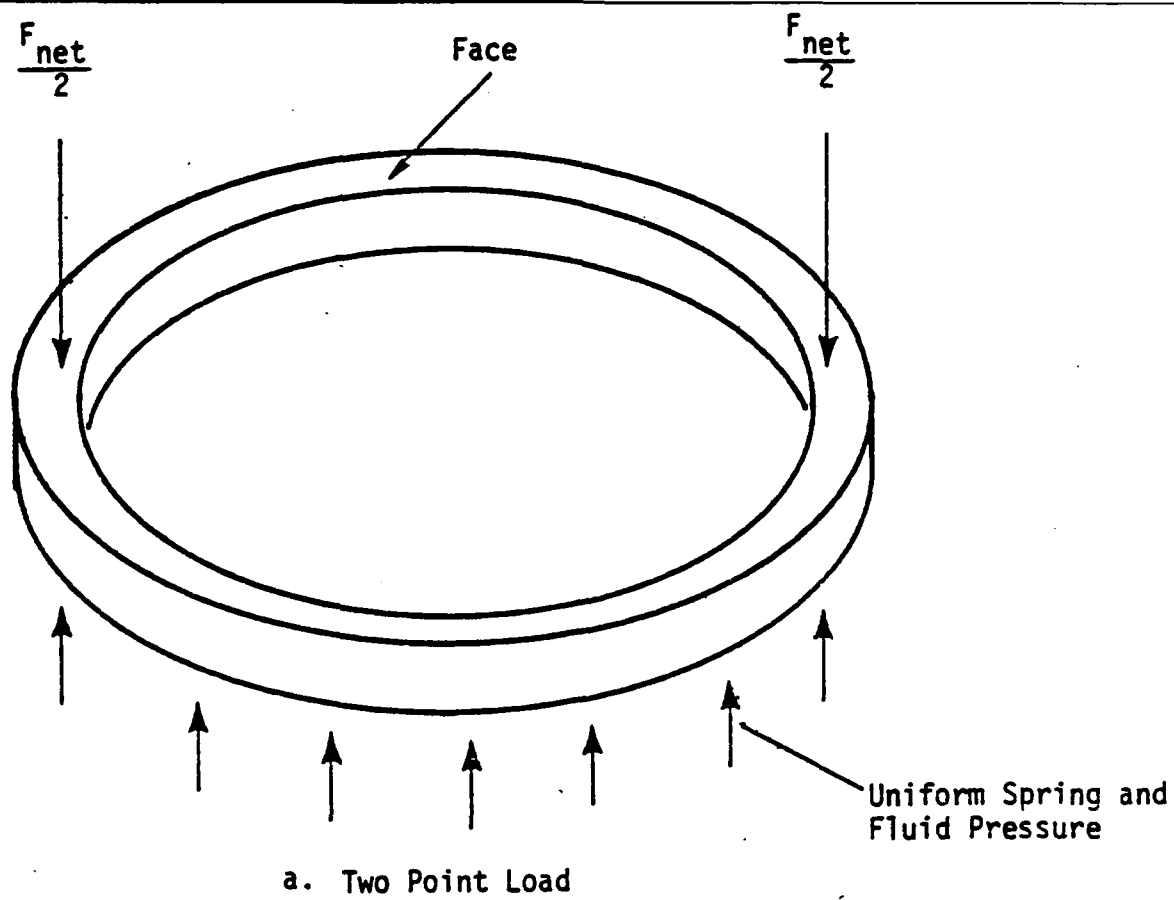


Figure 5-7. Flattening Load Distribution.

Deflection formulas have been derived for both cases using the methods for ring deflection developed in Reference [39]. For case a in Figure 4, there are n equally spaced concentrated loads V_{ey} , balanced by a uniformly distributed load. Case b in Figure 5-7 is for a sinusoidal load balanced by a uniform load. Both cases are given in Table 5-2.

Before the flattening calculations can be made, the net load in the above formulas must be considered. Only a fraction of the total load at the seal faces can act to flatten the seal faces. Assuming for the sake of simplification that the seal faces are neither significantly divergent or convergent or that such effects as caused by face taper average out to zero, the hydrostatic fluid pressure distribution across the face is linear and remains so regardless of film thickness. Therefore with respect to hydrostatic fluid pressure load support of the faces, the seal has no axial stiffness, and given the previous assumption, no circumferential variations in the hydrostatic fluid pressure load support can occur in spite of changes in film thickness. Therefore, the hydrostatic fluid pressure load support does not help to flatten out the seal faces. Only the mechanical or hydrodynamic part of the load support (that which must be provided to support the total load) can flatten a wave. To express this load in terms of a formula for an outside pressurized seal with zero inside pressure, the total load on the seal faces is

$$F_{\text{total}} = \pi(r_o^2 - r_i^2) (Bp_o + p_s) . \quad (5-9)$$

The hydrostatic load support is $1/2 p_o$ average. After subtracting,

$$F_{\text{net}} = \pi(r_o^2 - r_i^2) ((B - 0.5) p_o + p_s) . \quad (5-10)$$

Using the above formulas and Table 5-2, one can assess the flat-tenability of seal rings. Detailed studies using these simple formulas have been carried out in References [22], [41], and [42].

Two Ring Contact Model

For simple rings with constant cross sectional properties, no splits, and simple harmonic deflections, the face loading distribution required for continuous contact is relatively easily obtained as just shown. However, for more realistic conditions related to the cross section properties and more complex distortions, finding the correct distribution of face loading and seal gap becomes a very difficult problem. This development is presented in detail in Reference [21] and it will be summarized here.

The state of the art of predicting face loading in complex cases is illustrated by a report by Noell, Rippel, and Niemkiewicz of the Franklin Institute [43]. An evaluation of out-of-plane seal distortion caused by the nonuniformity of the joints in a split seal was made using finite elements applied to the rings and faces comprising the seal assembly. It is shown how a nonuniform cross section near the joints causes out-of-flatness of the faces. In the report the contact between the seal faces is modeled by springs where tensile stresses across the faces are allowed. Since such stresses cannot occur in

reality, the computed results do not predict how much and where the faces separate. Thus, the utility of the work to date is limited.

Theory

To define the problem of interest, one is trying to find the resultant gap, as a function of angular position, formed between two rings of arbitrary face profile and arbitrary geometry in a tangential direction, as they are loaded one upon the other by an arbitrary load. The variation in sealing gap results from two sources: 1) non-axisymmetric initial displacement (waviness) and 2) non-axisymmetric deflection due to non-axisymmetric loads (including face contact) or non-axisymmetric section properties. The loads arise from the contact itself, spring loads, hydrostatic loads, and drive force loads.

The problem can best be illustrated by looking at a projection of the circumferential centroidal axes of two rings as shown in Figure 5-8. Waviness of the surface may arise from production processes themselves or other types of distortion and creep. It is assumed that the centroidal axes initially have no distortion. As the two seal rings are brought together by the closing load, contact is initially established at three points. As the load is increased the contact points become regions and there may be any number of such regions. If the load is large enough and the waviness is not too steep, the gap may close around the entire seal and the contact pressure will be circumferentially variable. Alternatively, if the load is not large enough, regions of a seal gap will exist as shown in Figure 5-8.

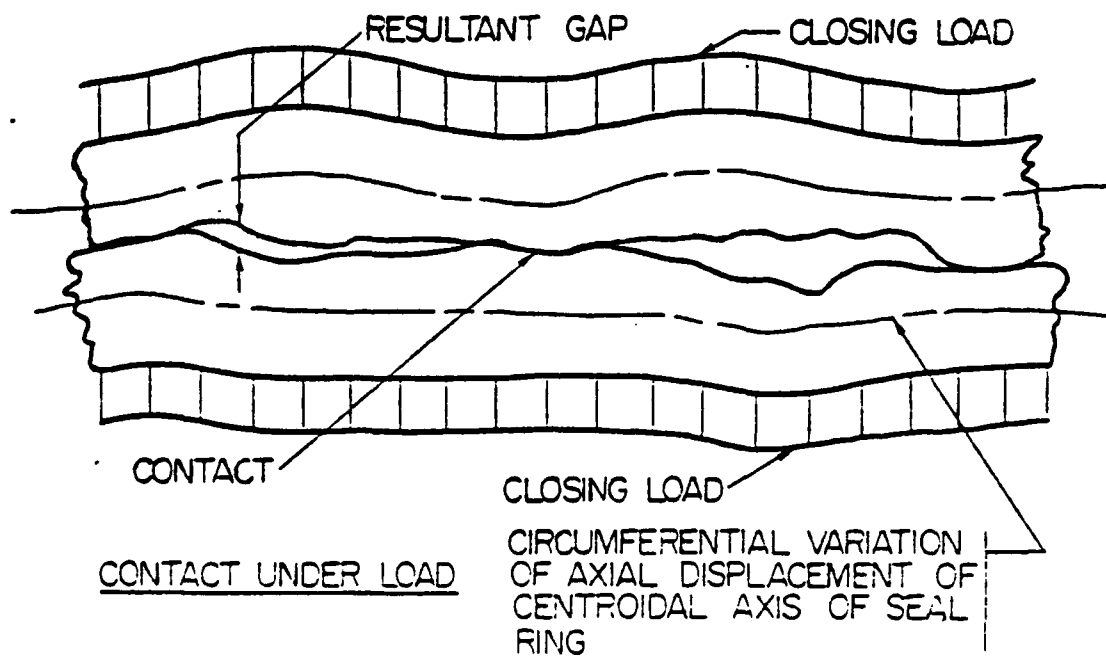
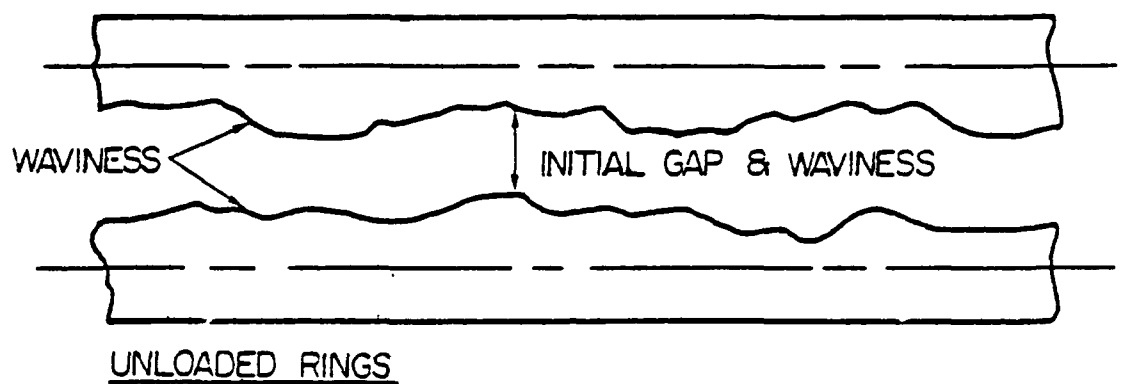


Figure 5-8. Contact of Two Rings.

There are two major features needed in the model to predict such contact/gaps. First, one needs a very general tool to predict deflections in arbitrarily loaded variable cross section rings. Second, as the two rings are brought together to contact, the pressure distribution must be predicted. The ring finite element described previously has been used to take care of the first feature. The second feature represents a specific type of contact problem. Such contact problems have been treated previously using finite element methods. These models are put together here to find the needed solutions.

The contact model assumes that contact can be represented by a system of linear springs as shown in Figure 5-9. Each spring represents localized deformation of the faces of the seal rings. If face deformation of the seal ring relative to its centroid is significant compared to the deformation of the centroidal axis itself (as it might be if one ring is carbon) then the values of k can be chosen based on an estimate of this stiffness. If the relative deformation is small then the k values may be chosen accordingly and they will not really influence the results as long as they are not chosen so large as to numerically dominate the stiffness matrix and cause errors.

Contact is defined as when the spring touches and is compressed. A gap results if the spring cannot touch due to either initial or elastic deformation due to load. Mathematically, when touching cannot occur, the spring constant is set to zero.

Figure 5-10 shows how the ring and contact models are combined. The figure represents the plane of one node in the finite element technique to be described. Note that each of the ring cross sections can

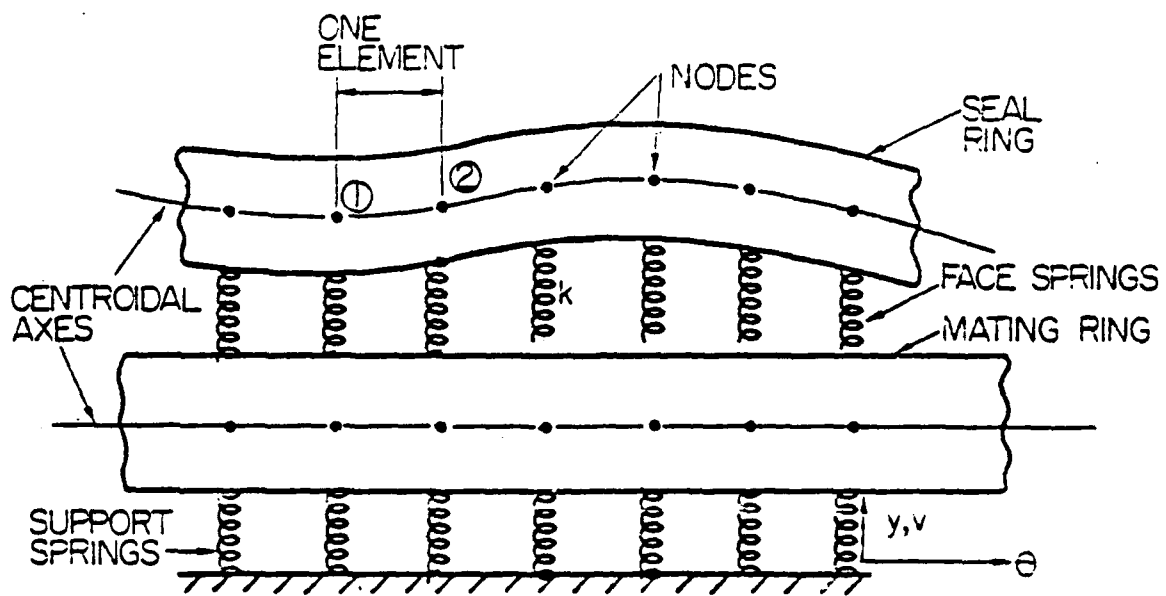


Figure 5-9. Contact model.

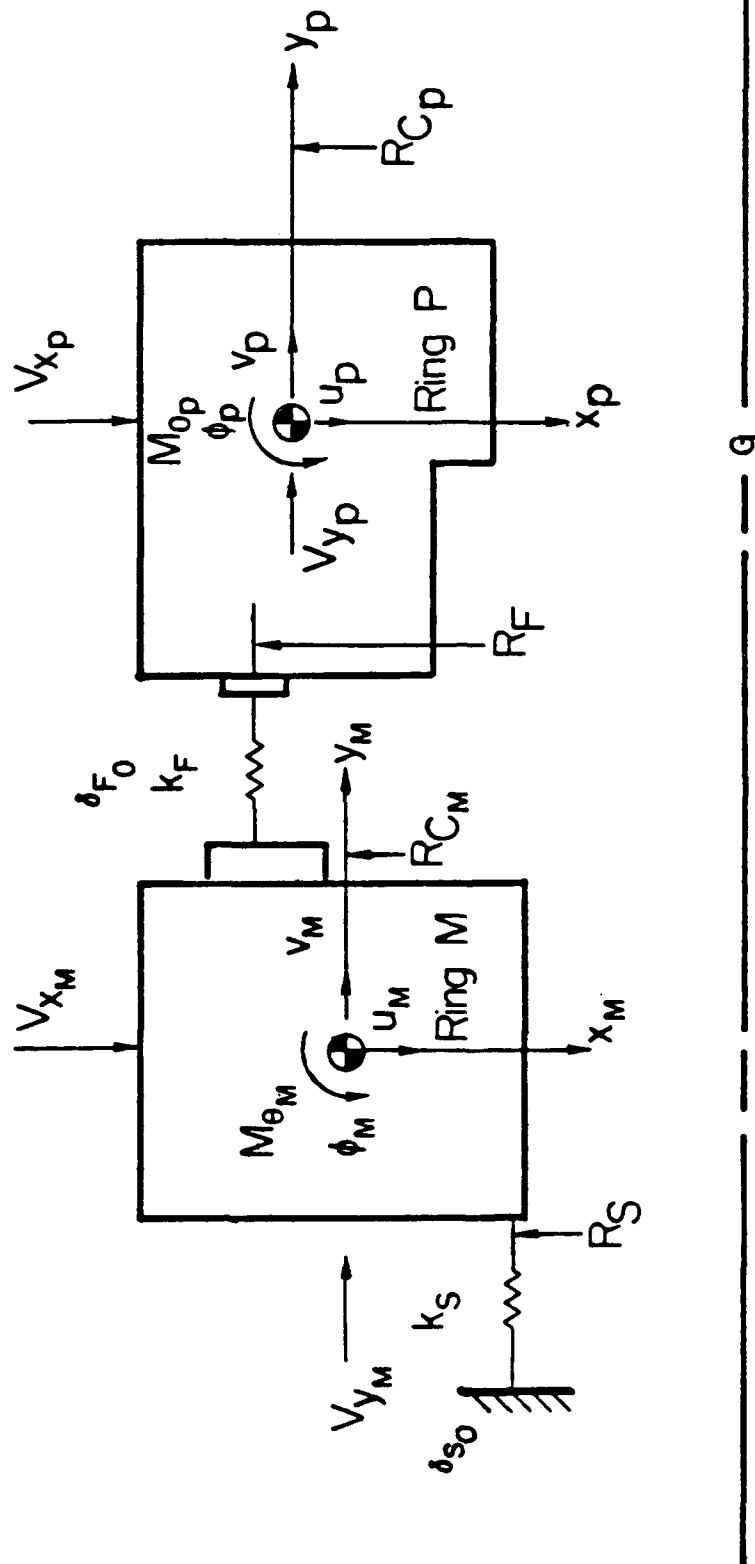


Figure 5-10. One Element of Contact Model.

be loaded by concentrated forces and moments. These are used to represent various external loadings. In an actual seal assembly these forces and moments would arise from fluid pressure, spring pressure, and possibly some drive force components. The contact of the two faces is represented by springs k_F which have initial deflection δ_{FO} (to later account for initial waviness). The left-hand ring is anchored by a series of springs k_S (with initial deflection δ_{SO}) which can be chosen to represent a very conformable support like an O-ring by making k_S small or a very rigid support such as a machined shoulder with an initial waviness on it by providing δ_{SO} and making k_S large.

To formulate the two ring deflection problem as a finite element problem, one must find the total energy of the system. The strain energy arises from that in the two rings, the face contact springs and the support springs. The potential energy is that due to the applied loads. Two ring finite elements plus two one-half face springs and two one-half support springs will be defined as one element in this combined problem (see Figure 5-10). It is readily shown that the strain energy plus the potential energy for such an element is given by

$$\begin{aligned}
 U_e = & \frac{1}{2} [\delta_p]^T [K_{RP}] [\delta_p] + \frac{1}{2} [\delta_M]^T [K_{RM}] [\delta_M] \\
 & + \frac{1}{2} \frac{k_{F1}}{2} [v_{M1} - v_{P1} - \phi_{M1}(R_F - R_{CM}) + \phi_{P1}(R_F - R_{CP}) - \delta_{FO1}]^2 \\
 & + \frac{1}{2} \frac{k_{F2}}{2} [v_{M2} - v_{P2} - \phi_{M2}(R_F - R_{CM}) + \phi_{P2}(R_F - R_{CP}) - \delta_{FO2}]^2 \\
 & + \frac{1}{2} \frac{k_{S1}}{2} [-v_{M1} - \phi_{M1}(R_{CM} - R_S) - \delta_{SO1}]^2
 \end{aligned}$$

$$\begin{aligned}
& + \frac{1}{2} \frac{k_{S2}}{2} [-v_{M2} - \phi_{M2}(R_{CM} - R_S) - \delta_{S02}]^2 \\
& - [\delta_P]^T [F_P] - [\delta_M]^T [F_M] .
\end{aligned} \tag{5-11}$$

The matrices $[K_{RP}]$ and $[K_{RM}]$ are as defined by Equation (5-6) for the two different rings. $[\delta_P]$ and $[\delta_M]$ are defined by Equation (5-1) and represent a total of 24 displacements associated with each of these elements. The four expressions with the k coefficient are recognized as being the potential energy of the springs.

Equation (5-17) can be rewritten in the following form

$$U_e = \frac{1}{2} [\delta_e]^T [K_e] [\delta_e] - [\delta_e]^T [F_e] - [C_e] \tag{5-12}$$

where $[F_e]$ includes initial deflection terms as well as loads and $[C_e]$ is a constant due to initial deflection. Subscript e indicates that the matrices are for one element. For the above form, Equations (5-13) and (5-14) (next page) give δ_e and F_e .

The stiffness matrix $[K_e]$ is somewhat cumbersome but can be represented as follows:

$$\begin{aligned}
 [s_e] = & \begin{bmatrix} u_{p1} \\ v_{p1} \\ w_{p1} \\ \frac{1}{R} \frac{dv}{d\theta} \Big|_{p1} \\ \psi_{p1} \\ \phi_{p1} \\ u_{M1} \\ v_{M1} \\ w_{M1} \\ \frac{1}{R} \frac{dv}{d\theta} \Big|_{M1} \\ \psi_{M1} \\ \phi_{M1} \\ u_{p2} \\ v_{p2} \\ w_{p2} \\ \frac{1}{R} \frac{dv}{d\theta} \Big|_{p2} \\ \psi_{p2} \\ \phi_{p2} \\ u_{M2} \\ v_{M2} \\ w_{M2} \\ \frac{1}{R} \frac{dv}{d\theta} \Big|_{M2} \\ \psi_{M2} \\ \phi_{M2} \end{bmatrix} \quad (5-13) \quad [F_e] = \begin{bmatrix} -v_{xp1} \\ -v_{yp1} + \frac{k_{F1} \delta_{F01}}{2} \\ -N_{\theta p1} \\ M_{xp1} \\ -M_{yp1} \\ -M_{\theta p1} - \frac{k_{F1} \delta_{F01}}{2} (R_F - R_{CP}) \\ -v_{xM1} \\ -v_{yM1} - \frac{k_{F1} \delta_{F01}}{2} + \frac{k_{S1} \delta_{S01}}{2} \\ -N_{\theta M1} \\ M_{xM1} \\ -M_{yM1} \\ -M_{\theta M1} + \frac{k_{F1} \delta_{F01}}{2} (R_F - R_{CM}) + \frac{k_{S1} \delta_{S01}}{2} (R_{CM} - R_S) \\ v_{xp2} \\ v_{yp2} + \frac{k_{F2} \delta_{F02}}{2} \\ N_{\theta p2} \\ -M_{xp2} \\ M_{yp2} \\ M_{\theta p2} - \frac{k_{F2} \delta_{F02}}{2} (R_F - R_{CP}) \\ v_{xM2} \\ v_{yM2} - \frac{k_{F2} \delta_{F02}}{2} + \frac{k_{S2} \delta_{S02}}{2} \\ N_{\theta M2} \\ -M_{xM2} \\ M_{yM2} \\ M_{\theta M2} + \frac{k_{F2} \delta_{F02}}{2} (R_F - R_{CM}) + \frac{k_{S2} \delta_{S02}}{2} (R_{CM} - R_C) \end{bmatrix} \quad (5-14)
 \end{aligned}$$

$$\begin{bmatrix}
 \begin{matrix} K_{RP1,1} & & \\ & \ddots & \\ & & K_{RP6,6} \end{matrix} & 0 & \begin{matrix} K_{RP1,7} & & \\ & \ddots & \\ & & K_{RP6,12} \end{matrix} & 0 \\
 0 & \begin{matrix} K_{RM1,1} & & \\ & \ddots & \\ & & K_{RM6,6} \end{matrix} & 0 & \begin{matrix} K_{RM1,7} & & \\ & \ddots & \\ & & K_{RM6,12} \end{matrix} \\
 \begin{matrix} K_{RP7,1} & & \\ & \ddots & \\ & & K_{RP12,6} \end{matrix} & 0 & \begin{matrix} K_{RP7,7} & & \\ & \ddots & \\ & & K_{RP12,12} \end{matrix} & 0 \\
 0 & \begin{matrix} K_{RM7,1} & & \\ & \ddots & \\ & & K_{RM12,6} \end{matrix} & 0 & \begin{matrix} K_{RM7,7} & & \\ & \ddots & \\ & & K_{RM12,12} \end{matrix}
 \end{bmatrix}$$

(5-15)

Each nonzero 6 x 6 submatrix is taken from part of the appropriate 12 x 12 ring stiffness matrix defined by Equation (5-6). One 12 x 12 is defined for ring P and one for ring M. The elements of the P ring stiffness matrix and the M ring stiffness matrix must have the same length θ to properly combine above.

The coupling terms which derive from Equation (5-11) must now be added to the 24 x 24 stiffness matrix. Since these terms are sparse, they are summarized by Equation (5-16) (next page) as additions to the stiffness matrix above.

Thus

$$[K_e] = [Eq. (5-15)] + [Eq. (5-16)] . \quad (5-17)$$

i, j	2	6	8	12	14	18	20	24
2	$\frac{k_{F1}}{2}$	$-\frac{k_{F1}}{2} (R_F - R_{CP})$	$-\frac{k_{F1}}{2}$	$\frac{k_{F1}}{2} (R_F - R_{CM})$	0	0	0	0
6	$-\frac{k_{F1}}{2} (R_F - R_{CP})$	$\frac{k_{F1}}{2} (R_F - R_{CP})^2$	$\frac{k_{F1}}{2} (R_F - R_{CP})$	$-\frac{k_{F1}}{2} (R_F - R_{CM})$ $\times (R_F - R_{CP})$	0	0	0	0
8	$-\frac{k_{F1}}{2}$	$\frac{k_{F1}}{2} (R_F - R_{CP})$	$\frac{k_{F1} + k_{S1}}{2}$	$-\frac{k_{F1}}{2} (R_F - R_{CM})$ $+\frac{k_{S1}}{2} (R_{CM} - R_S)$	0	0	0	0
12	$\frac{k_{F1}}{2} (R_F - R_{CM})$	$-\frac{k_{F1}}{2} (R_F - R_{CP})$ $\times (R_F - R_{CP})$	$-\frac{k_{F1}}{2} (R_F - R_{CM})$ $+\frac{k_{S1}}{2} (R_{CM} - R_S)$	$\frac{k_{F1}}{2} (R_F - R_{CM})^2$ $+\frac{k_{S1}}{2} (R_{CM} - R_S)^2$	0	0	0	0
14	0	0	0	0	$\frac{k_{F2}}{2}$	$-\frac{k_{F2}}{2} (R_F - R_{CP})$	$-\frac{k_{F2}}{2}$	$\frac{k_{F2}}{2} (R_F - R_{CM})$
18	0	0	0	0	$-\frac{k_{F2}}{2} (R_F - R_{CP})$	$\frac{k_{F2}}{2} (R_F - R_{CP})^2$	$\frac{k_{F2}}{2} (R_F - R_{CP})$	$-\frac{k_{F2}}{2} (R_F - R_{CM})$ $\times (R_F - R_{CP})$
20	0	0	0	0	$-\frac{k_{F2}}{2}$	$\frac{k_{F2}}{2} (R_F - R_{CP})$	$\frac{k_{F2} + k_{S2}}{2}$	$-\frac{k_{F2}}{2} (R_F - R_{CM})$ $+\frac{k_{S2}}{2} (R_{CM} - R_S)$
24	0	0	0	0	$\frac{k_{F2}}{2} (R_F - R_{CM})$	$-\frac{k_{F2}}{2} (R_F - R_{CP})$ $\times (R_F - R_{CP})$	$-\frac{k_{F2}}{2} (R_F - R_{CM})$ $+\frac{k_{S2}}{2} (R_{CM} - R_S)$	$\frac{k_{F2}}{2} (R_F - R_{CM})^2$ $+\frac{k_{S2}}{2} (R_{CM} - R_S)^2$

Equation (5-16)

It must be recognized that the moments and forces will combine in assembly such that the total term at each node is equivalent to the external load at each node.

Using conventional summation techniques to assemble the elements and the principle of minimum potential energy, it can be shown that

$$\left(\sum_{e=1}^N [K_e] \right) \{\delta\} = \sum_{e=1}^N [F_e] \quad (5-18)$$

defines the linear problem to be solved. $\{\delta\}$ represents the equilibrium displacements and becomes a matrix of 12 N unknowns for a ring with N nodes and N elements. To minimize bandwidth, $[K_e]$ was assembled around the ring using an alternating numbering scheme.

Boundary conditions must be supplied for Equation (5-18) to prevent rigid body motion. For the system as shown in Figure 5-10, the needed boundary conditions are

$$\begin{aligned} u_{M1} &= 0 & u_{P1} &= 0 \\ w_{M1} &= 0 & w_{P1} &= 0 \\ w_{MN/2} &= 0 & w_{PN/2} &= 0 \end{aligned} \quad (5-19)$$

These constraints keep the rings from rotating about their own axes and eliminate rigid body motion in the θ -x plane.

Method

Before considering solutions to the above set of equations, it is important to summarize the assumptions in this model. They are:

- 1) Ring finite elements are used, so the seal parts must be able to be suitably modeled as such. Ring finite elements give only translation of the centroid and rotation of the cross section.
- 2) The ring finite element derivation assumes a compact section where the effects of warping and shear center offset are negligible. Small deflection theory is used. Deflections due to shear are neglected.
- 3) Seal face taper is not accounted for in the contact model. Contact is assumed to occur only at the mean face radius. Thus the analysis is most accurate for narrow-faced seals or where face taper deflection (ϕ in the model) is small relative to face axial translation (v in the model).
- 4) Contact is represented by a linear spring model which represents deformation of the surface relative to centroid. The model is one dimensional.

The method of solution used for the above problem is as follows:

- 1) Evaluate ring stiffness matrices K_p and K_M for actual sections using Equation (5-6).
- 2) Define all other input including the face springs, the support springs k_{Si} , and external loads $[F]$.
- 3) Assume at least three points on the faces touch. Such points were found by an iterative scheme to pick the three most likely points of contact under zero load pushing the rings together. Place springs k_{Fi} at these points with all other face springs set to zero. Depending on the type of support,

similar initial conditions can be established for the k_{Si} . For the cases studied the k_{Si} were all set to the same value thus representing a uniform support of the seal ring. Given the assumed contact and the waviness of the surface, then the δ_{FO_i} and δ_{SO_i} can be calculated and input.

- 4) The external load $[F]$ is divided into M increments so that one increment of load is given by

$$\frac{1}{m} [F] . \quad (5-20)$$

One such increment is input as the load vector on the right-hand side of Equation (5-18).

- 5) The element stiffness matrices $[k_e]$ are constructed. Note this requires adding appropriate terms from the two ring stiffness matrices from Equation (5-6) in the form of Equations (5-15) and (5-16). The element stiffness matrices, Equation (5-17), require complete information concerning k_{Fi} , k_{Si} , δ_{FO_i} , δ_{SO_i} . Thus, the element stiffness matrices must be reconstructed each time the k_{Fi} and k_{Si} are changed.
- 6) Assemble the element stiffness matrices as in Equation (5-18). Use assembly rules to minimize bandwidth and store in compact form.
- 7) Solve Equations (5-18) using Gauss elimination.
- 8) Examine the deflection at each point on the face:

$$\Delta_i = v_{Mi} - v_{Pi} - \phi_{Mi}(R_F - R_{CM}) + \phi_{Pi}(R_F - R_{CP}) . \quad (5-21)$$

Then

$$\text{if } \Delta_i \geq \delta_{FOi} \quad k_{Fi} = k_F , \quad (5-22)$$

$$\text{if } \Delta_i < \delta_{FOi} \quad k_{Fi} = 0 . \quad (5-23)$$

A similar procedure is followed for the support. This step places springs in place only where contact is occurring.

- 9) Add another increment to the load and repeat steps 4) through 8) m times.
- 10) Repeat steps 5) through 8) again without adding any more load (the full load is already applied). If the contact conditions, Equations (5-22) and (5-23), are the same at each node, a correct solution is obtained. If such cannot be achieved, increasing the value of m will usually achieve a satisfactory solution.

The use of the increment of load was found to be essential to obtain a converged, consistent solution. Other approaches were tried with no success.

The final result is the complete set of displacements (5-13) at each node. In addition one also gets the seal gap h .

$$\text{if } \Delta_i < \delta_{FOi} , \text{ then } h_i = \delta_{FOi} - \Delta_i , \quad (5-24)$$

$$\text{if } \Delta_i > \delta_{FOi} , \text{ then } h_i = 0 . \quad (5-25)$$

Experimental Checks

The algorithm detailing steps 1) through 10) has been checked in several ways. By a suitable choice of constraints one may solve any

individual ring problem. Several such problems were solved thus verifying the ring stiffness matrix and the assembly of matrices. Several problems were also solved to check out the spring support system. Again, these checks verified the modeling. The model has also been applied to two experimental situations, one a contrived seal-like configuration, discussed below, the other, a real face seal for which data was available discussed in a later section.

Figure 5-11 shows the details of two aluminum rings and a loading arrangement designed to cause the rings to separate at the contact face. The moment arm is attached at node 1 as shown. As the load is applied the faces separate starting at node 1. Additional load causes the separation to move around the ring. Separation will also occur at node 5.

Table 5-3 gives the details of the section properties for the tests. Using the computer program developed and using only eight nodes, the resulting gap was predicted and is shown in Table 5-3. Experimental results are shown in Table 5-3 for comparison. Considering the numerous assumptions in the model and the experimental error, agreement is considered good. Of particular note is the fact that theory and experiment agree that node 5 just barely lifts off at the given load. Thus, this test result provides some direct confirmation of the validity of the model.

The model is now applied to some important and practical problems. The numerical model itself is described in detail in Appendix C.

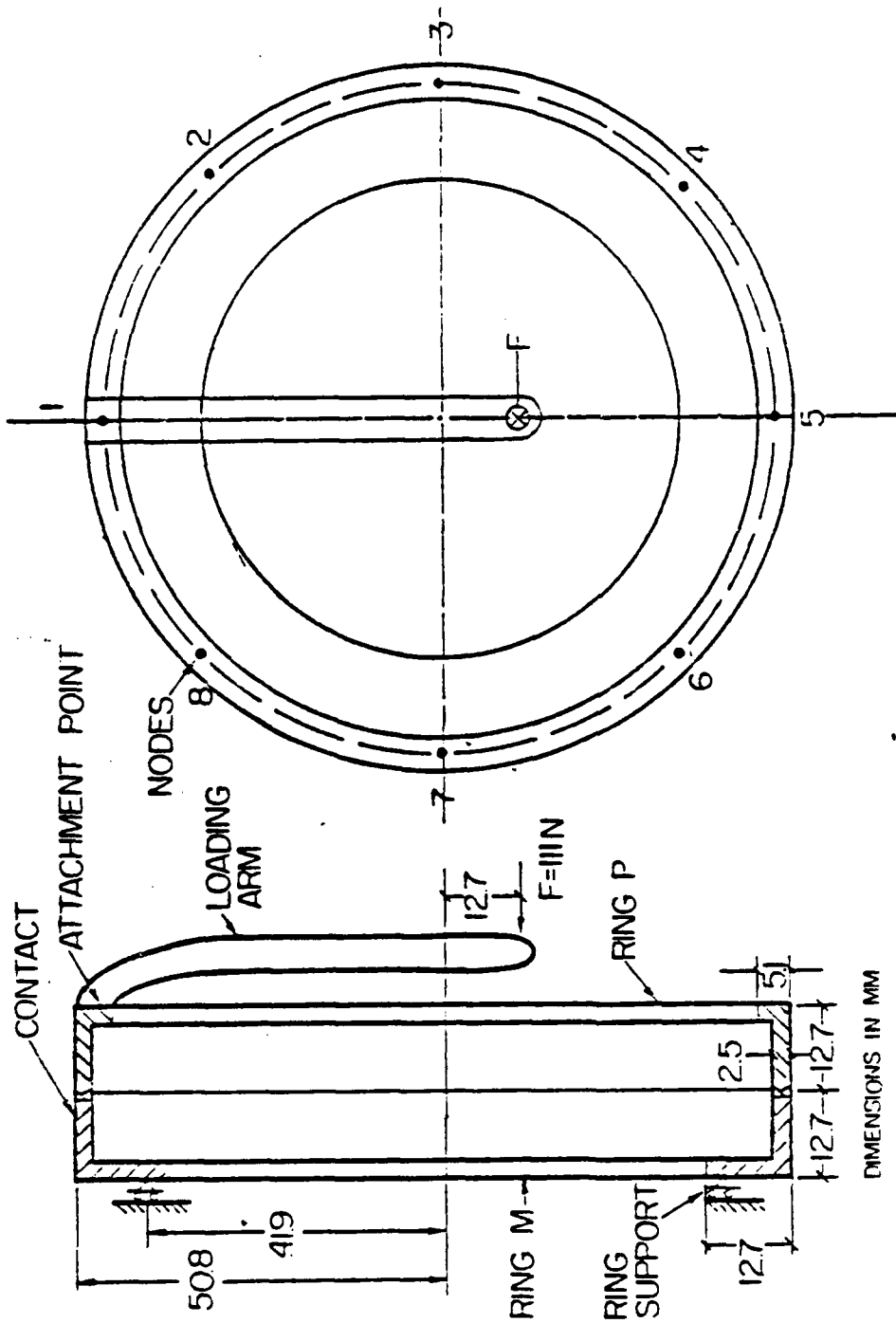


Figure 5-11. Two Contacting Rings with Eccentric Load.

TABLE 5-3

Two Contacting Rings with Eccentric Load

Section Properties (aluminum)		
	Ring M	Ring P
R_c	48.0 mm	50.4 mm
a	58.1 mm ²	38.7 mm ²
J_x	$8.03 \times 10^3 \text{ mm}^4$	$5.77 \times 10^2 \text{ mm}^4$
J_y	$8.85 \times 10^3 \text{ mm}^4$	$5.66 \times 10^1 \text{ mm}^4$
J_{xy}	$4.87 \times 10^2 \text{ mm}^4$	$7.18 \times 10^1 \text{ mm}^4$
J_θ	$1.25 \times 10^2 \text{ mm}^4$	$8.33 \times 10^1 \text{ mm}^4$

Results

Node	Predicted Gap (mm)	Measured Gap (mm)
1	0.65	0.51
2	0.29	0.20
3	0.0	0.0
4	0.0	0.0
5	0.04	0.04
6	0.0	0.0
7	0.0	0.0
8	0.29	0.20

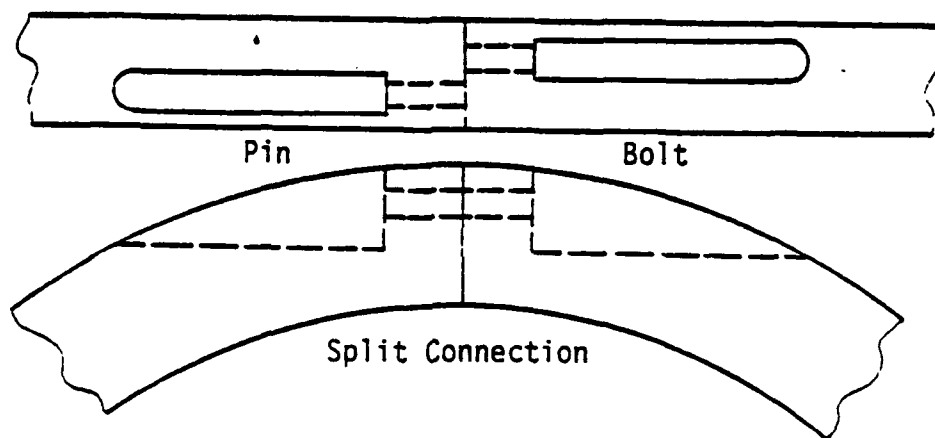
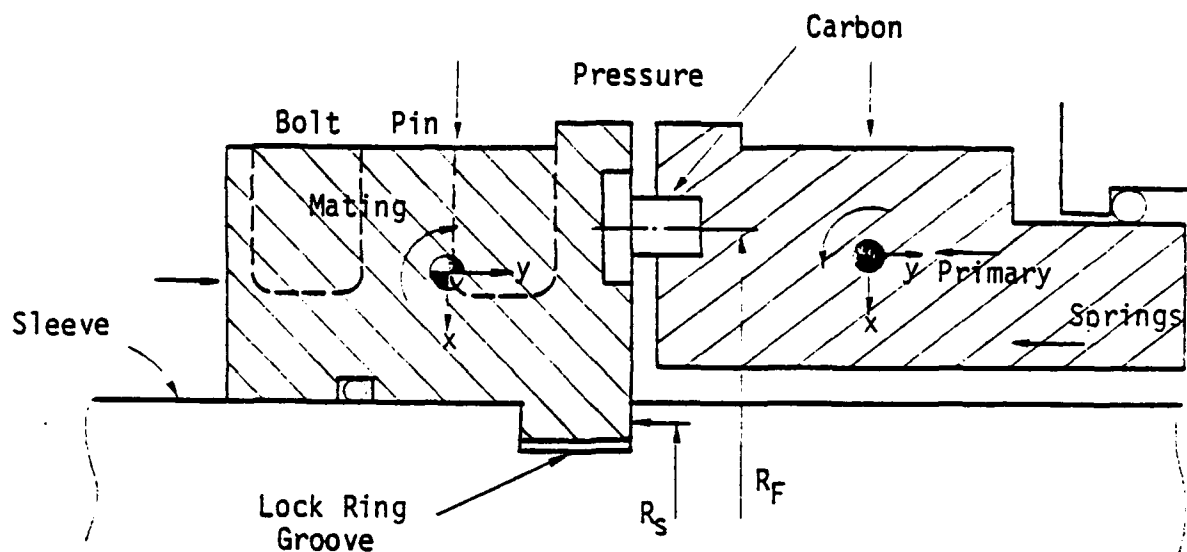
Two Ring Contact Model Studies--Submarine Seal

A representative submarine seal is shown in Figure 5-12. The primary ring floats but does not rotate. It contains a carbon insert wearing element. Axial hydrostatic pressure and spring pressure load the primary face onto the mating ring face. The radial hydrostatic pressure causes a tilt which causes a divergence of the seal as well as a compressive hoop stress and radial deformation. The primary ring seals against the mating ring which also has a wear insert. The mating ring rotates with the shaft sleeve. The radial pressure creates a divergent rotation in the mating ring as well. The axial pressure load is far greater than that applied by the primary ring so that the mating ring is forced to contact the edge of the lock ring groove. This means that the mating ring will, under large pressure, tend to take on the shape of the locking groove.

Both the primary and secondary rings are made of monel (recent designs use Inconel 625). Both rings are split as shown. The two halves are bolted together using a taper pin and a bolt. Thus the cross section is reduced on both rings in two regions.

Section properties for 688 class submarine seals are shown to give some idea of the size of the seal rings and the cross section. The J_0 values used are not exact but taken from an approximate formula. I values have been used instead of J values. The figure shows how the bolt or pin cutout reduces the section properties (at the point of maximum cutout).

To illustrate the importance of seal gap and leakage in a submarine seal, Figure 5-13 shows the profile of an unloaded primary ring



Section Properties

Mating	Mating at Pin Cutout	Mating at Bolt Cutout	Primary
$R = 13.08$	13.808	13.808	$R = 13.844$ in
$I_x = 14.15$	12.92	10.91	$I_x = 38.18$ in ⁴
$I_y = 9.76$	7.31	7.31	$I_y = 6.70$ in ⁴
$J_\theta = 24.40$	8.11	9.00	$J_\theta = 21.45$ in ⁴
$I_{xy} = 1.73$	2.93	-0.33	$I_{xy} = 2.99$ in ⁴
$a = 11.52$	9.00	9.00	$a = 13.30$ in ²

Figure 5-12. Submarine Shaft Seal.

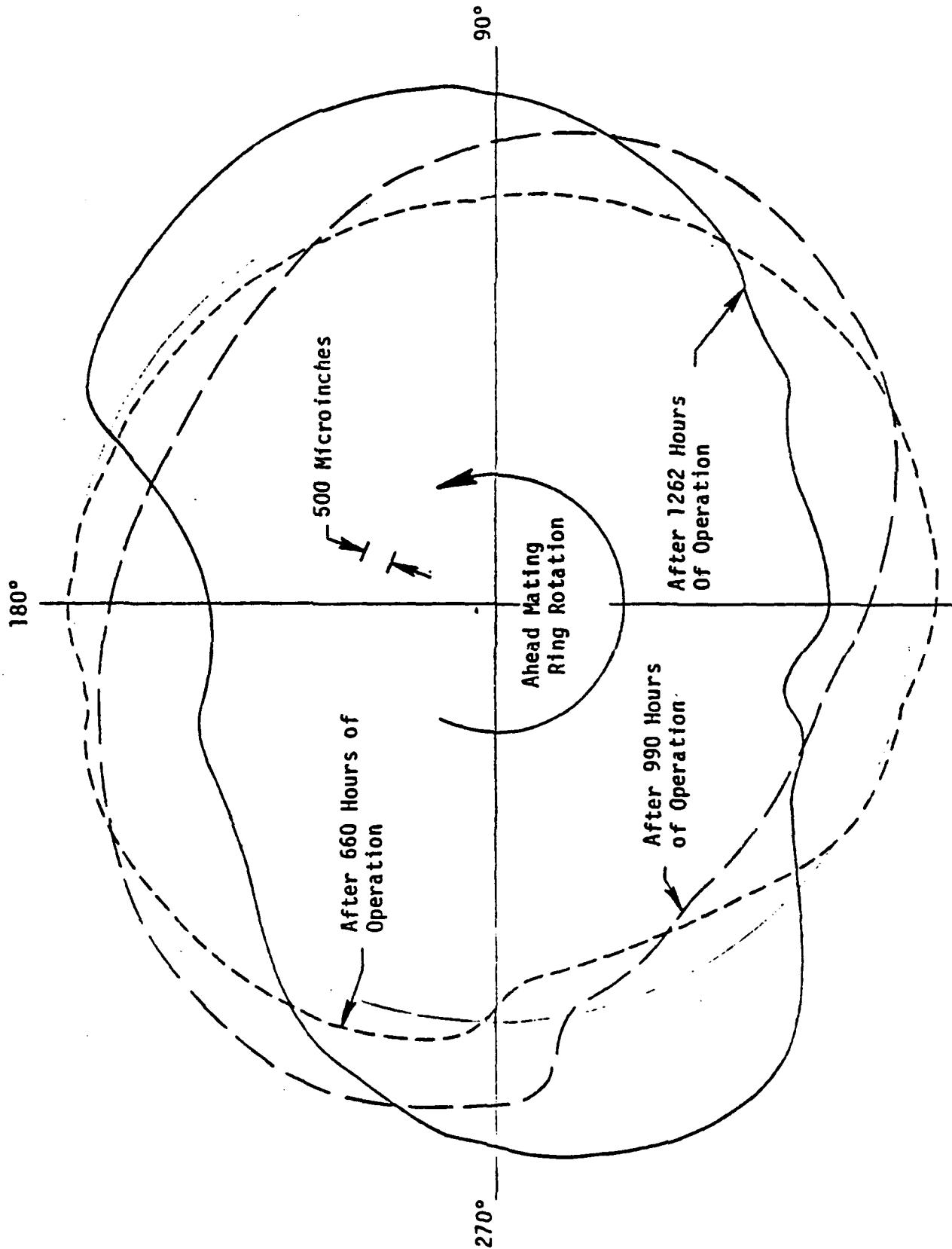


Figure 5-13. Seal Ring Face Flatness.

that develops after a period of operation. This seal was leaking badly (10 GPM). Table 5-4 shows that high leakage would be expected from this seal unless the gap shown in Figure 5-13 closes up on contact with the opposing face.

The two ring contact model is ideally suited to study such problems in submarine seals. Several different kinds of studies have been made using the program to try to assess submarine seal characteristics in regard to waviness and conformability. Table 5-5 summarizes the results of these studies which are now discussed in detail.

Seal Gap Caused by Bolted Joints

The bolted joints in a seal causes a non-axisymmetric stiffness. This variable stiffness coupled with the large moment associated with the radial hydrostatic component of pressure load can cause the faces to go out of flat. This condition was assessed using the two ring program for the seal described in Figure 5-12. A number of assumptions were made for this analysis.

- 1) Primary ring in perfectly flat and axisymmetric effects are eliminated for this analysis.
- 2) Mating ring is perfectly flat before loading. Reduced sections due to bolt/pin cutouts are included.
- 3) Mating ring is modeled in two ways: with joint as stiff as ring itself (assuming a large preload and a perfect joint) and assuming that the bolts are just snugged tight and stretch elastically as the joint comes open.
- 4) Lock ring groove is perfectly flat.

TABLE 5-4

Significance of Seal Gap

Q^*	Q	h_{avg}	h_2 equivalent
<u>(gpm)</u>	<u>ml/min</u>	<u>(μin)</u>	<u>(μin)</u>
10.0	37850	882	650
1.0	3785	409	301
0.1	378	190	140
0.01	38	88	65
0.001	4	41	30

*For 688 size and 100 percent pressures
at 60°F

TABLE 5-5
Submarine Seal Conformability Studies

Study	Case	Cube mean of face* Gap (μ in)	Results
1 Effect of seal ring cutout and bolted joint (688 mating ring)	Tight joint Loose joint	0 111	Does not open Seal opens due to moment on joint
2 Conformability of maximum flattenable waviness (688 primary)	$h_2 = 795 \mu$ in	0	Waviness was completely flattened.
3 Effect of lock ring groove stiffness on mating ring conformability (688)	$h_2 = 1590 \mu$ in KS = 10 x KF KS = KF KS = 0.1 x KF	122 91 1	Leakage = 112 ml/min Leakage = 46 ml/min Leakage = 0 ml/min
4 Effect of a 0.001" offset in seal face	Present design Backfit design	100 99	Gap = 15° (3.62") Gap = 10° (2.41") 60 ml/min
5 Effect of shim in lock ring groove (Trident)	2" x 0.003" shim 2" x 0.006" shim	186 529	Leakage = 460 ml/min (Experimental = 150 ml/min [43]) Leakage = 10,900 ml/min (Experimental = 5000 ml/min [43])
6 Effect of 2nd harmonic wave in lock ring groove (Trident)	500 μ in amplitude 650 μ in amplitude 1000 μ in amplitude	0 1 83	No gaps in face Gaps just barely develop Large gaps in face

*The cube mean rather than the simple average is the proper value of h to use when estimating the leakage.

- 5) Face stiffness is based on face carbon. Support stiffness is taken as 10X face stiffness to represent a very rigid (metal on metal) support.
- 6) A variable node spacing is used to account for the joints and the cutouts.
- 7) The mating ring does not separate from the lock ring groove (later studies show it does and this causes even more gap than predicted here at the faces).

Using the stiff joint assumption the seal does not open under load although the face loading does become variable because of the non-axisymmetry. If the bolts are just tightened, the faces do separate. This occurs because the large moment opens the joint slightly and distorts it out of flat. Because of the high rigidity of the rings, this distortion cannot be flattened back out by the deflection of the rings. Figure 5-14 shows the actual gap which results. The negative deflection portion represents the axial deflection of the carbon insert. Table 5-5 shows that the effective gap is 111 μ in., enough to cause a significant leak.

Now the actual joint does not correspond to either of these extremes. The preload on the bolts is high but not high enough to insure that the seal joint will behave the same way as the parent material. Loose joints are also not realistic. In fact, it is thought that the stiffness characteristics of a joint such as this are non-linear because the fraction of the joint separated increases with

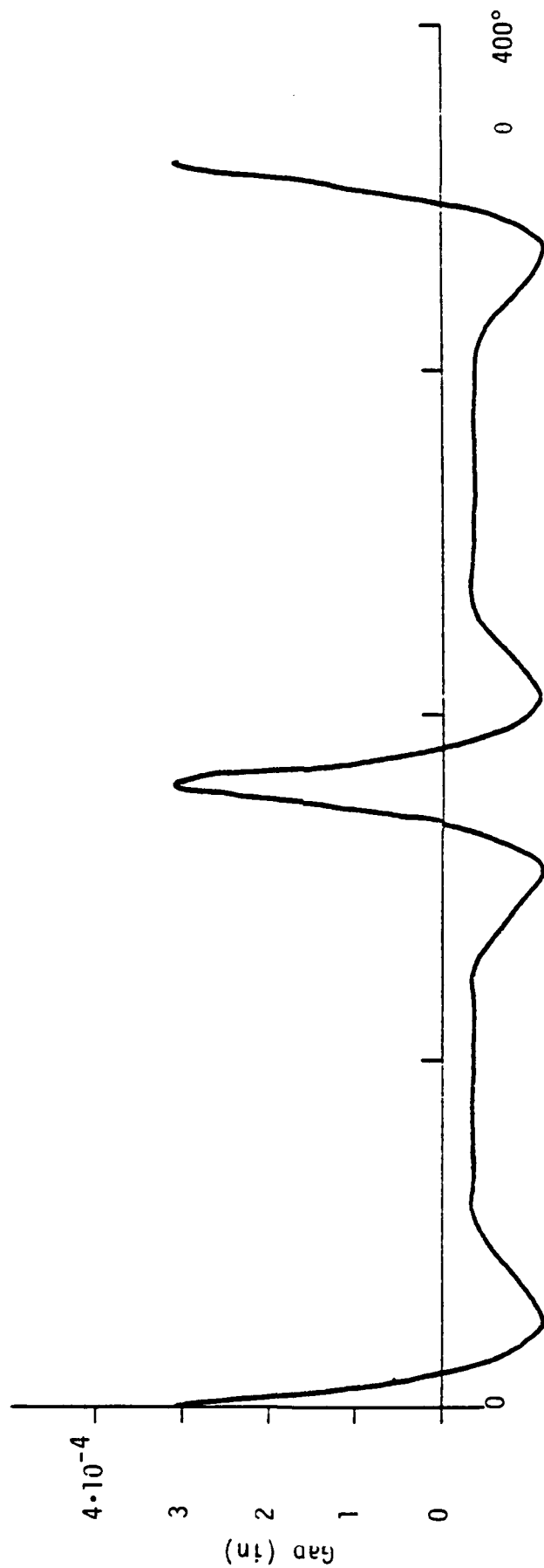


Figure 5-14. Seal Gap due to Joints.

increasing load. Thus a more accurate joint model is needed to accurately predict seal gap resulting from such non-axisymmetric effects. This subject is discussed in detail in a later section of this report.

The results presented here must be taken as a first estimate of the effects of non-axisymmetric stiffness on a seal. The important thing that these preliminary results show is that the joints in concert with the large moment tend to cause the seal to open, and this may in fact, at the very least, aggravate seal leakage problems. The most effective remedy is to eliminate the large moment on the ring. This will eliminate a large fraction of the out of flatness caused by joints.

Conformability of Primary Ring to a Wave

A second harmonic wave of 795 $\mu\text{in.}$ amplitude was introduced on the face of the primary ring (see Table 5-5). Both rings were modeled axisymmetrically. The mating ring was made infinitely stiff. The results from the contact model showed that the seal faces just touched all around. The contact pressure dropped to zero at two points indicating that the seal faces were just about to separate. This case was run to verify the program result using a simple formula prediction as described previously. Reference [42] details these formula predictions, this one being for the 688 at 88 percent balance. (A simple study showed that the maximum flattenable waviness for the mating ring was 1270 $\mu\text{in.}$, however the mating ring is stiffened by the lock ring groove.)

The significance of second harmonic waviness flattenability is discussed in detail in Reference [22]. Maximum flattenable second harmonic waviness serves as an indicator of the relative conformability of seals. It is known that seals that leak a lot often have a worn in waviness much greater than that which can be flattened out--thus the high leakage. The relative conformability serves as a good guide for designing seals with adequate conformability, but there are many other complications to the conformability question as the next few studies will show.

Lock Ring Groove Studies

The mating ring in current submarine seals is held in place on the shaft sleeve by a lock ring which fits a groove in the sleeve as shown in Figure 5-12. Pressure caused forces on the ring press the mating ring onto the face of the groove. There are two consequences. First the mating ring tends to take the shape of the groove, i.e., groove waviness is reflected in the mating ring. Second, the groove acts to stiffen the mating so that its conformability is not fully utilized. These effects are studied here.

Table 5-5, study 3, shows the stiffening effect of the lock ring groove. A 1590 μ in. amplitudal wave is imposed on the seal faces. According to the maximum flattenable wave for the primary plus the mating ring, this amount of wave should be easily flattened out. The first case of study 3 shows that with a rigid lock ring groove (the actual case) a gap develops large enough to cause significant leakage (Figure 5-15). Thus the groove acts to stiffen the mating ring so that

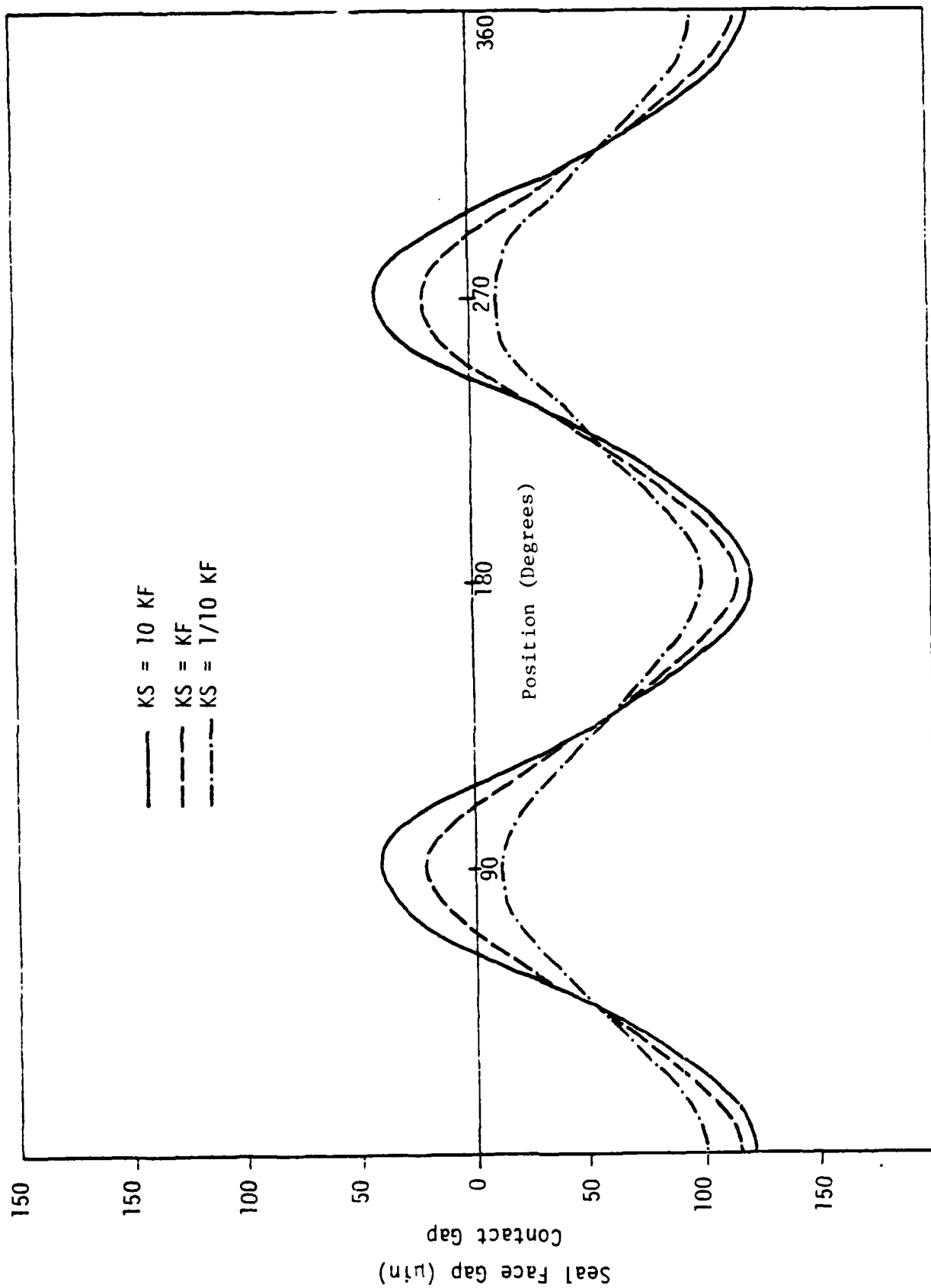


Figure 5-15. Conformability of Seals with Differing Support Stiffnesses.

it does not readily conform to the mating ring in the presence of a wave on the faces.

If the groove itself is allowed to be compliant such as by allowing the ring to press against a rubber cushion instead of metal, case 3 of study 3 shows that the face gap disappears (Figure 5-15). Thus this study shows that the lock ring groove reduces the conformability of the seal. Little is gained by having the mating ring be compliant.

The solution to the problem is to allow the mating ring to become nearly free floating so that it is not forced to take a shape from some other part. Such designs have to be tried and do show an improvement [43].

Study 6 in Table 5-5 shows that effect of putting a second harmonic wave directly into a lock ring groove. This study assumes the faces as originally flat and the rings are axisymmetric. As the amplitude of the lock ring waviness gets to around 0.001 in., the seal opens and cannot be closed any longer by the conformability of the faces. Figure 5-16 shows the results.

The solution to this problem is the same as above. The pressure load on the mating ring must be relieved so that the mating ring does not take its shape from another part.

Study 5 shows the effect of placing a 2 in. long shim in an otherwise flat lock ring groove. The model shows that the seal opens and this indicates in another way the previous point that seal leakage may be caused by lock ring out of flatness. Figure 5-17 shows these results. Interestingly this case was also run as a physical experiment

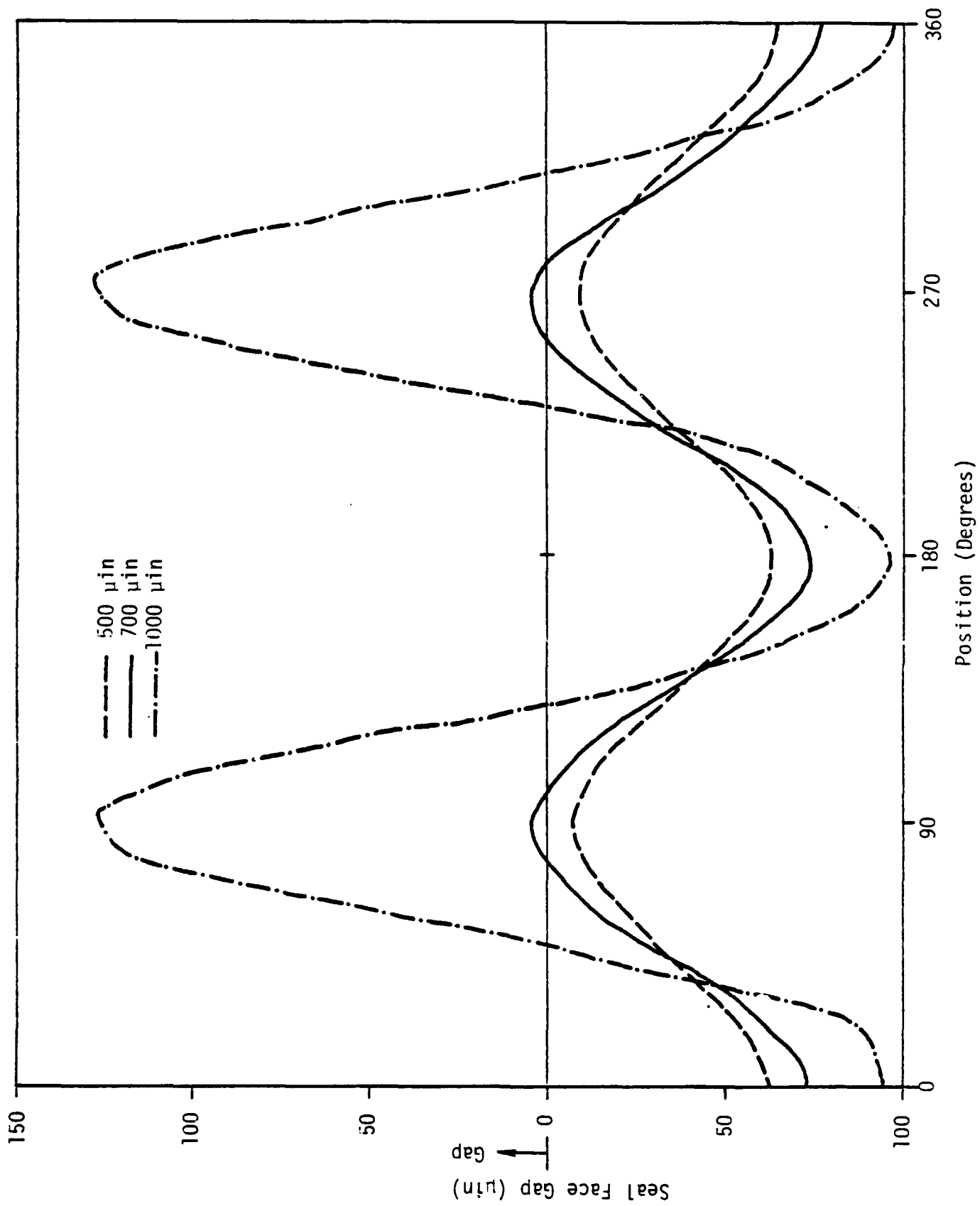


Figure 5-16. Effect of 2nd Harmonic Wave in Lock Ring Groove.

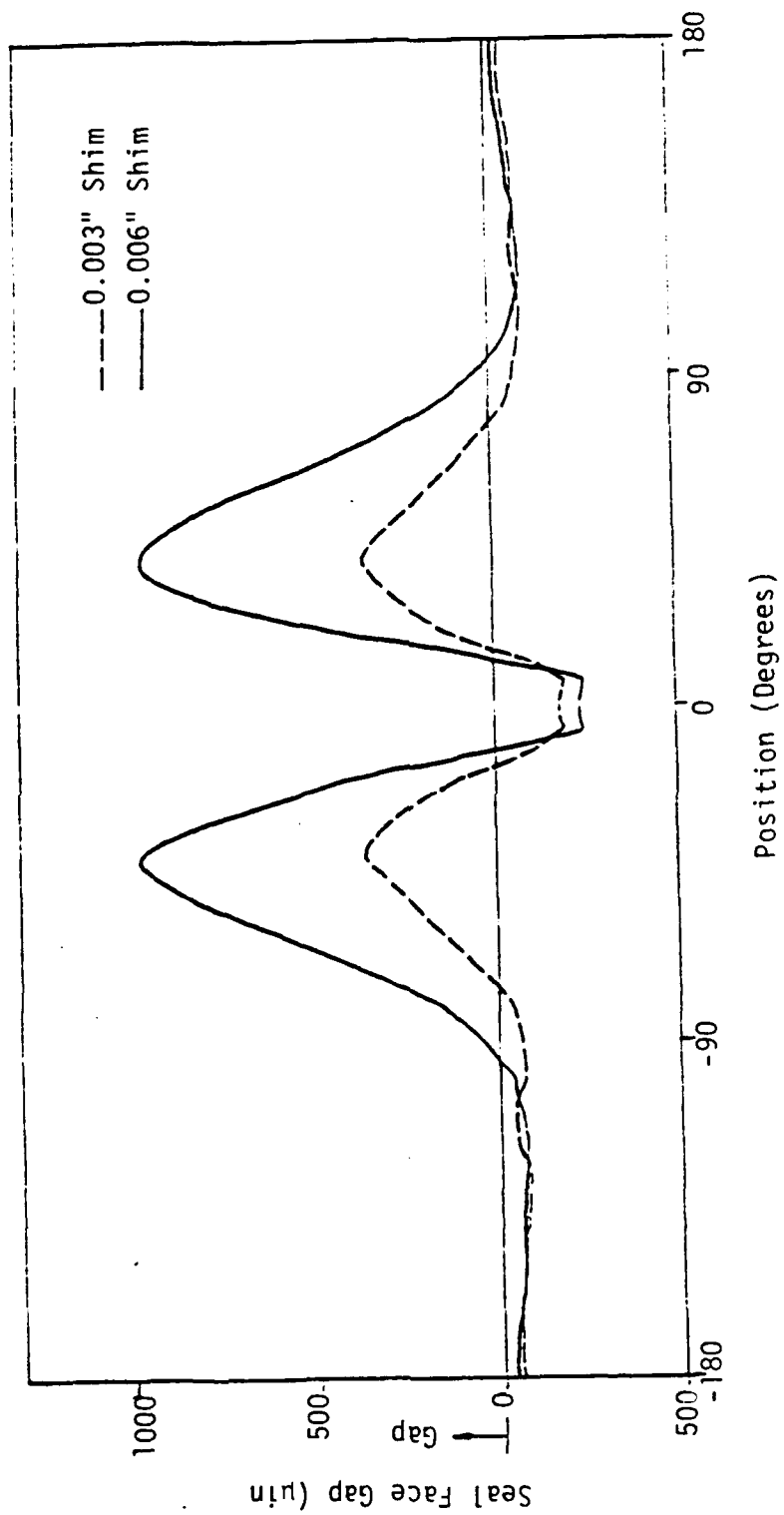


Figure 5-17. Effect of Shim in Lock Ring Groove

on a full scale seal. The measured leakage is shown. The model predicts similar results.

Face Offset

Study 4 shows the effect of an offset in the seal face as might occur if one seal segment shifts relative to the adjacent segment as has been known to occur in some tests. Clearly such a gap cannot be completely closed. Figure 5-18 shows the model's prediction of how the submarine seal will respond to a 0.001 in. face offset. The figure shows that there will be a large deformation of the carbon segment sticking up but about a 0.0005 in. gap will remain. The length of the gap is about 15 degrees before it completely closes. The cube mean gap value shows that a significant leakage will occur.

The solid curve in Figure 5-18 is for a seal design which is many times more compliant than the original design. The cube mean gap remains essentially the same and the length of the gap is reduced to 10 degrees.

The conclusion to be reached here are that submarine face seals cannot readily comply to segment offsets by bending of the rings even if the rings are quite flexible. Offsets such as these must either be minimized by design in the first place or must be complied with by mounting the segments themselves in a compliant manner such that these ends can have a relative movement to eliminate the offset.

The two ring contact model has been shown to be very useful in making such studies. The overall conclusions made from these studies are summarized in Chapter 9.

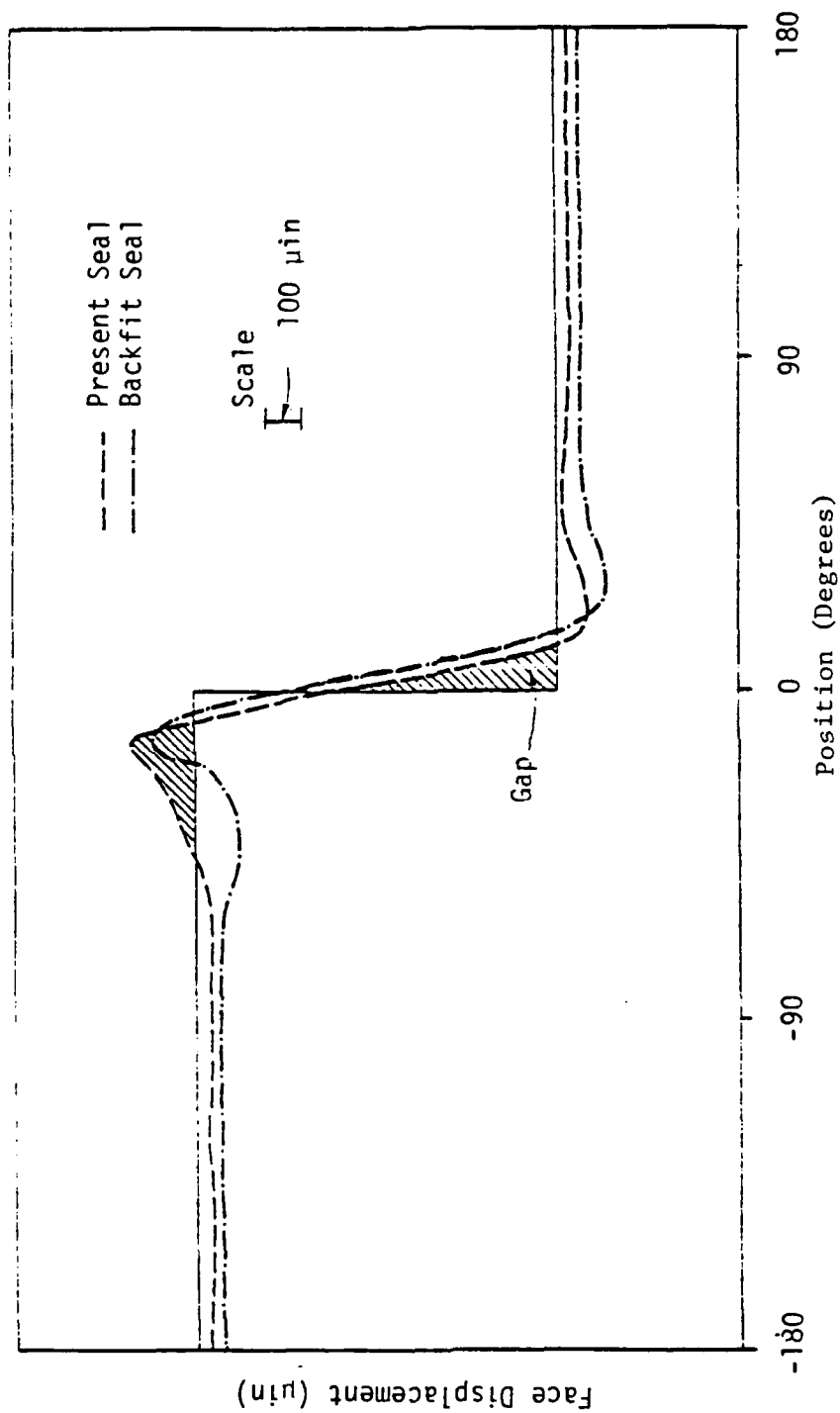


Figure 5-18. Effect of Face Offset.

Two Ring Contact Model-Magnetic Seal

The two ring contact model was also used to help understand leakage problems in some small magnetic seals used in Naval devices. The magnetic seal is shown in Figure 5-19. The magnet attracts the metal holder and pulls the carbon toward it. Waviness measurements of these seals were taken so that seal profiles were known. Leakage measurements as a function of angular positions were also taken. The complete details of this program are reported in Reference [41].

The two ring contact program was applied to the magnetic seal. Details of this application are contained in Appendix C. Figure 5-20 shows a specific example. The original measured profiles are shown. The net gap before deflection is shown. Then after the seal rings are loaded and bend toward each other, the net gap shown by the dashed curve results. Note that considerable flattening out occurs and that there are three regions of contact.

Leakage for the net gap was computed and then the two profiles were rotated relative to each other and the above solutions repeated. A different net gap and profile resulted. This procedure was repeated to establish a leakage versus relative position curve as shown in Figure 5-21. These results are compared to experimental results in the same figure. Agreement is considered to be reasonable since the model totally ignores radial effects (assumes the gap is parallel across the seal). These results show that the two ring contact model is a useful tool for understanding conformability problems in seals. Further details are given in Reference [41].

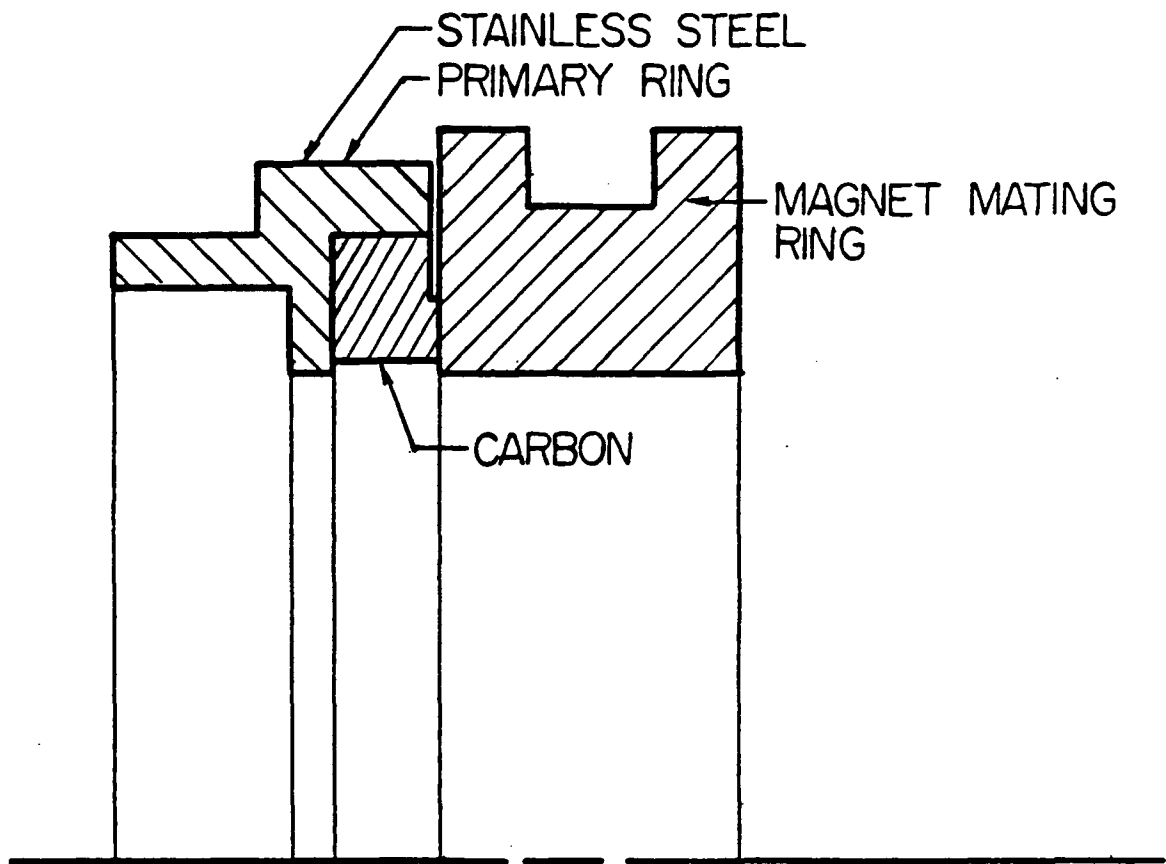


Figure 5-19. Magnetic Seal.

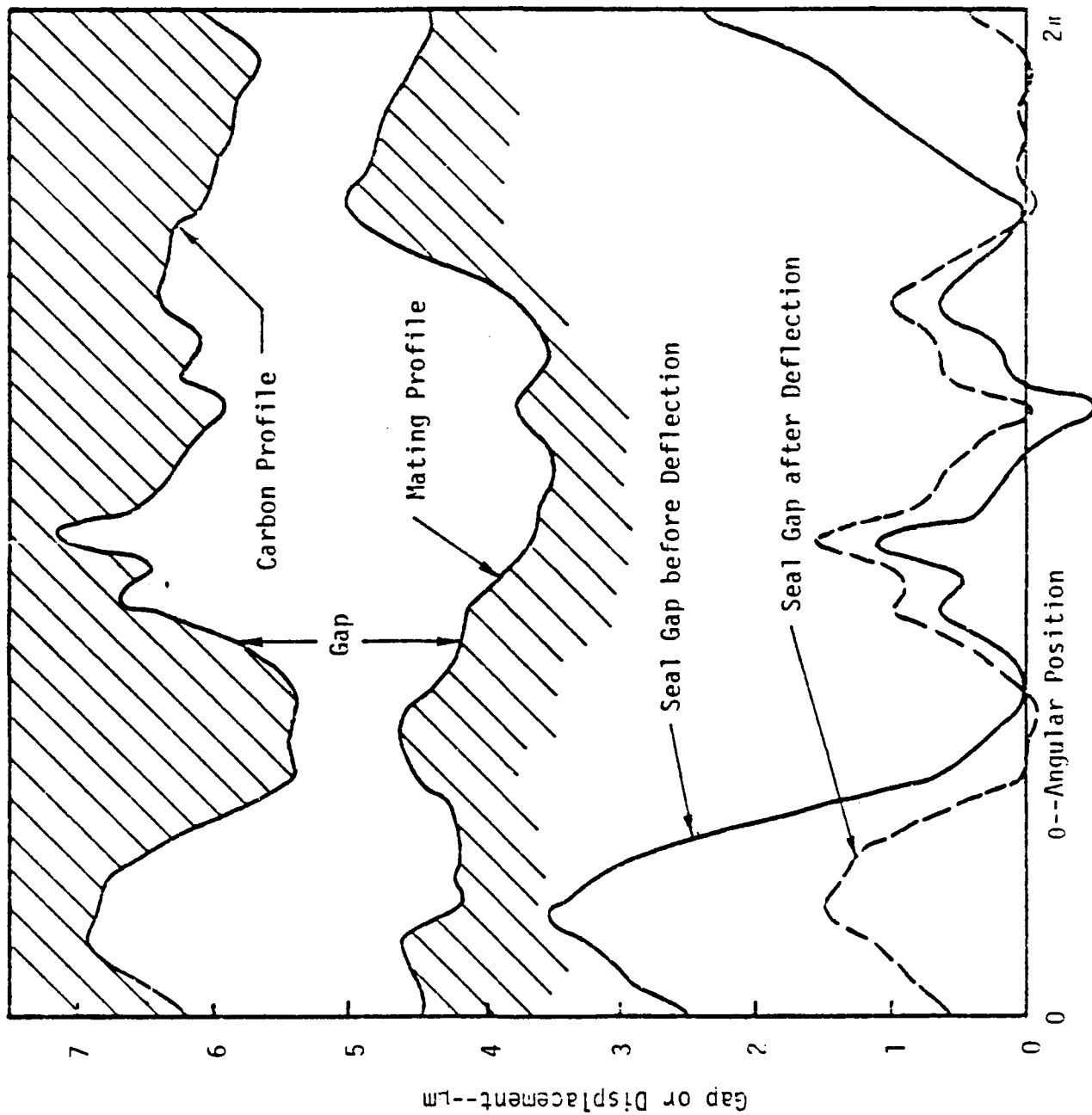


Figure 5-20. Surface Profiles and Seal Gap.

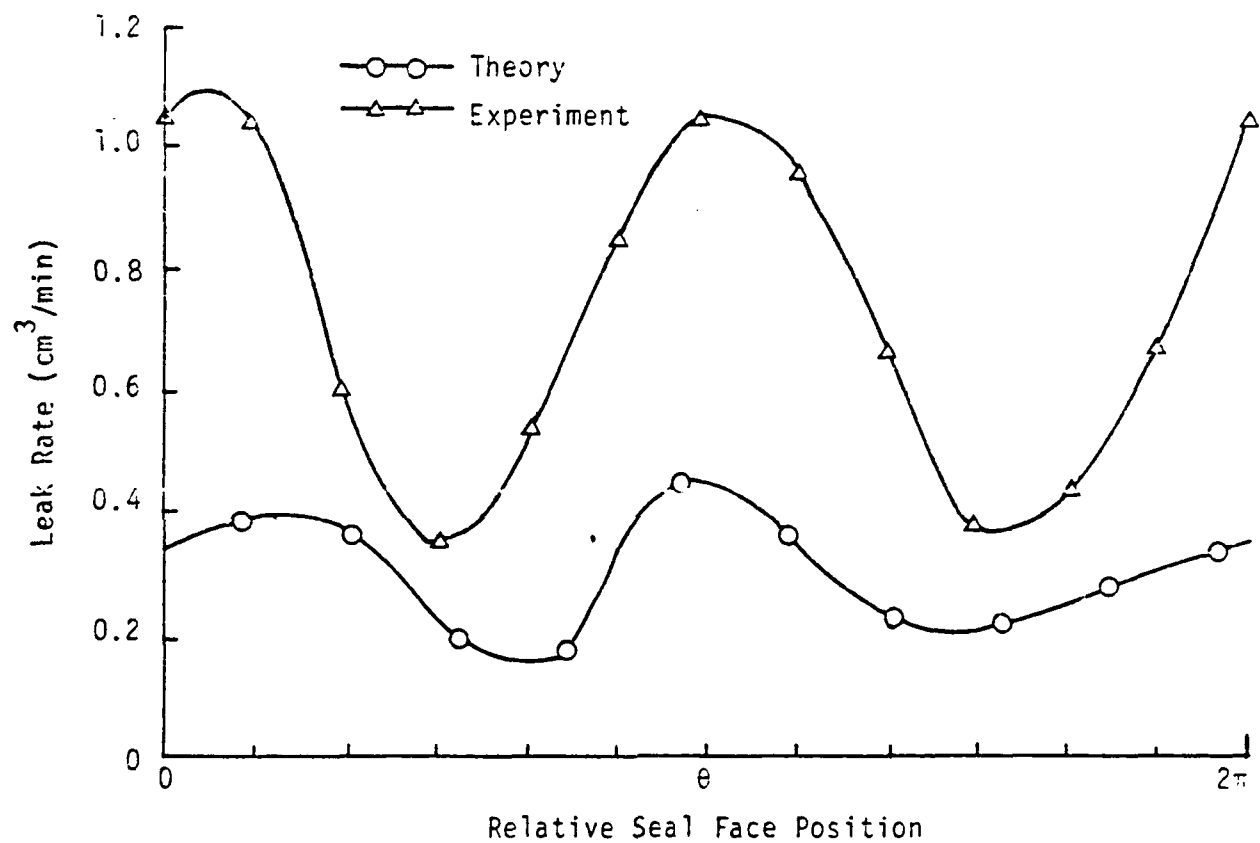


Figure 5-21. Comparison of Experimental to Theoretical Leakage.

Nonlinear Joint Model

As discussed previously, there is some question as to the proper way to model the bolted joints in the rings for the purpose of solving deflection and contact problems. Clearly assuming the joint behaves like the rest of the ring is incorrect. The loose bolt assumption is also incorrect. However, these two approaches represent cases which are readily modeled and have been used accordingly as a starting point.

A more precise definition of joint deflections versus forces is needed. If this relationship is known, it can readily be put into the ring finite element program of interest so that proper joint deflection is accounted for. As a part of this research program, it was decided that some studies should be undertaken to define these relationships so that proper seal ring deflection modeling could be performed.

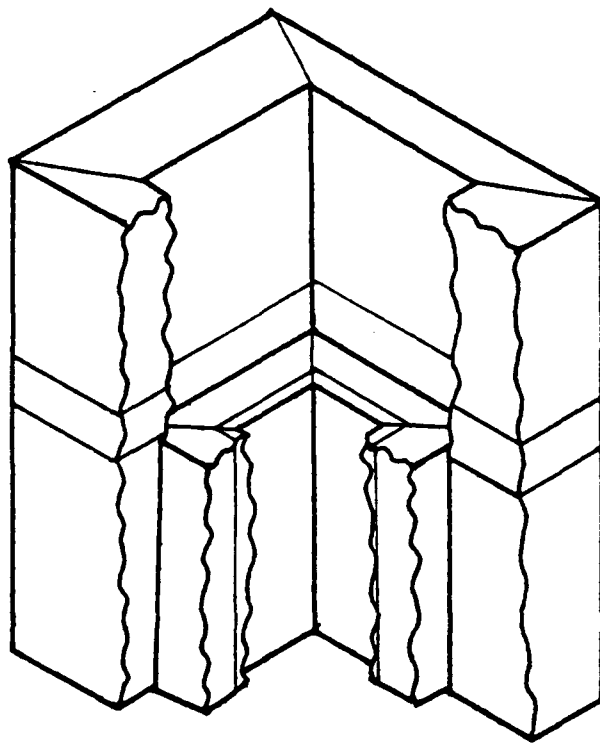
A literature search revealed that the moment-rotation characteristics of several different types of bolted joints have been investigated. Bose [44] and Krishnamurthy [45] investigated the moment-rotation characteristics of bolted connections commonly used in steel structures. Their work indicates that the response of top-angle, tee-stub, and end-plate connections to moments is nonlinear. Bose [44] reports the results of experiments performed on the three types of connections and shows that the joint stiffnesses decrease with increased loading. Krishnamurthy developed a two-dimensional finite element code to analyze these joints.

Tsutsumi, Ito, and Masuko [46] performed bending experiments on two types of bolted joints used in machine tools. Again, their results show that the joint stiffnesses decrease with increasing moment. Sawa, Maruyama, and Edamoto [47] examined the distribution of contact pressure in a joint with a tap bolt both analytically and experimentally. They showed that the pressure distribution depends on the fastener depth and is not constant across the interface.

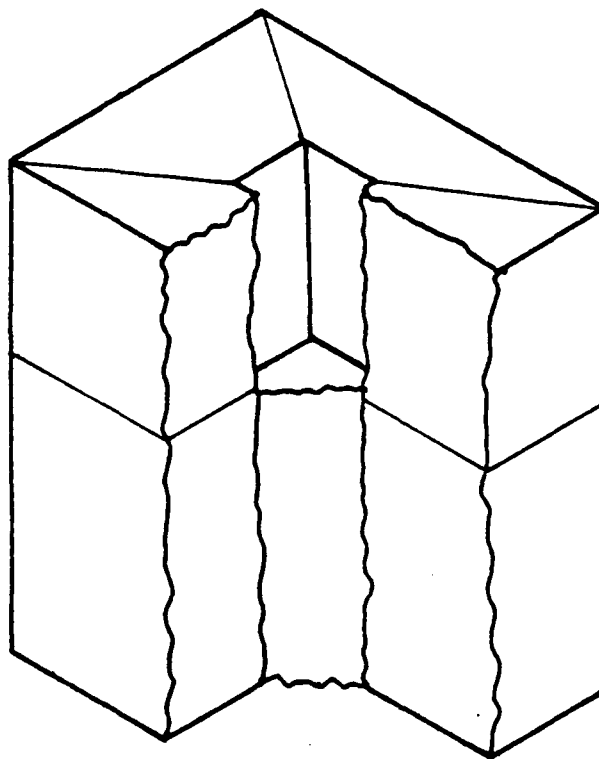
These papers suggest that the moment-rotation characteristics of tap bolted joints, as in the seal rings, are nonlinear. They do not, however, provide such a solution to the actual problem of interest.

A preliminary finite element numerical model of bolted joint similar to that in a seal ring has been developed. It consists of two rectangular blocks bolted together. This model is broken into a coarse finite element mesh of 32 elements and 88 nodes (Figure 5-22). The joint itself is modeled using three-dimensional, 8-node brick elements using the finite element code ADINA. The interfaces between the joint halves and between the bolt head and the joint are modeled using truss elements. These elements act as springs and allow contact between the joint halves and the bolt and also simulate the bolt pretension. Initially, their stiffnesses are large compared to the stiffnesses of the brick elements.

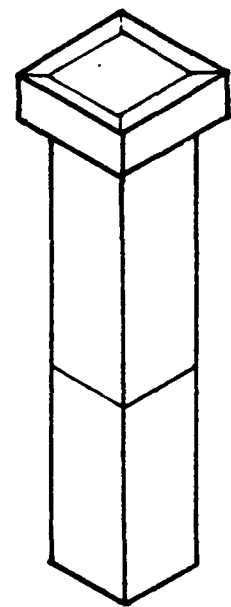
A stepwise increasing moment is applied to the joint. As forces in the springs become tensile, the stiffnesses of those particular elements are lowered to simulate the joint opening. The results (Figure 5-23) show that the joint's response is linear until it begins to open. As the load increases past that point, the joint stiffness



TOP (20 Elements)



BOTTOM (5 Elements)



BOLT (7 Elements)

Figure 5-22. Finite Element Mesh.

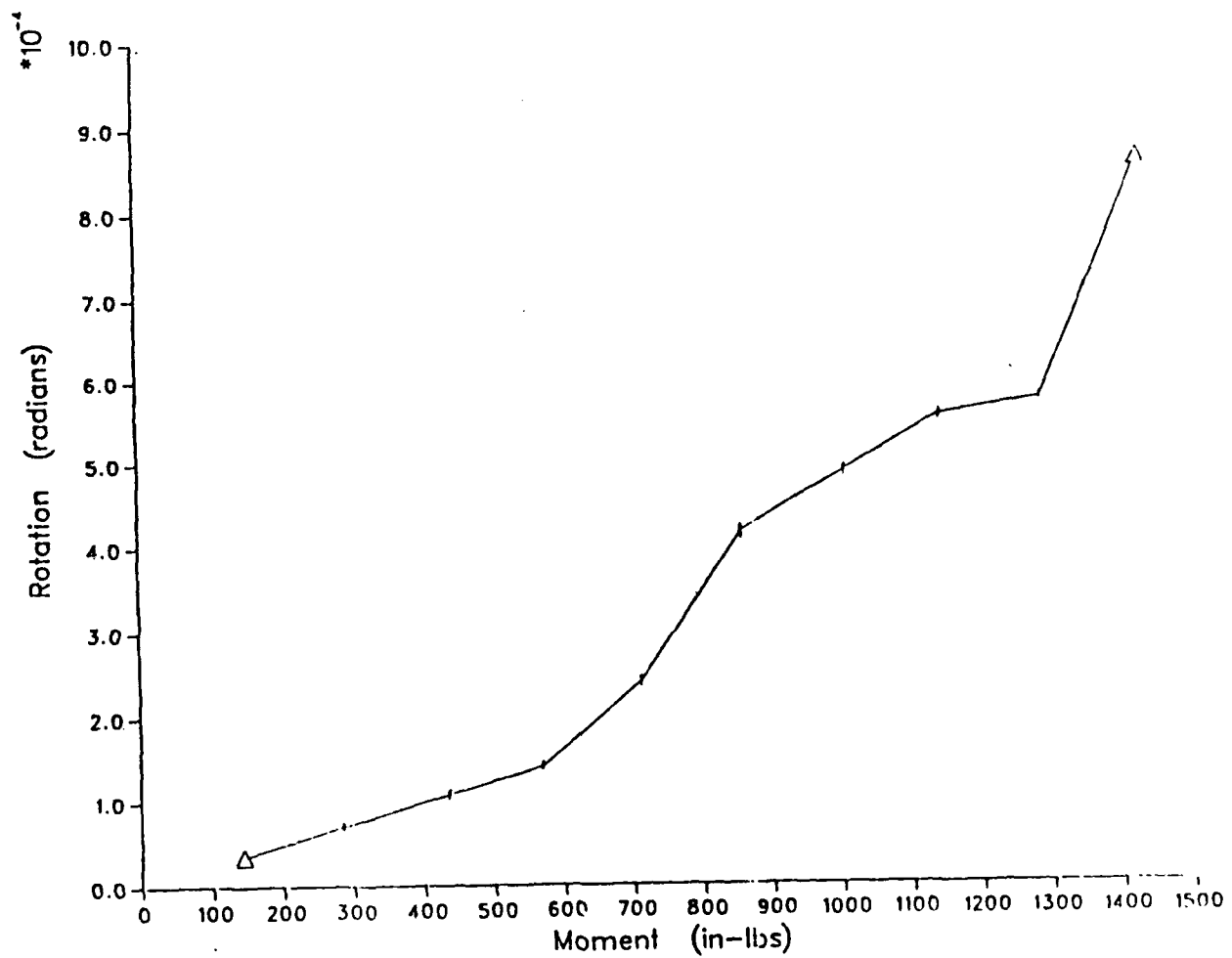


Figure 5-23. Moment-rotation Curve.

decreases giving a nonlinear response. The lack of smoothness in the curve in Figure 5-23 is due to the coarseness of the mesh, the large load increments and/or the placement of the applied loads.

It is proposed to further investigate the moment-rotation characteristics of the joint by greatly refining the mesh. A mesh generating program will be written and used to produce the refined element mesh. It will place the nodes and develop element connectivity. The program will allow the mesh size to be changed easily and will format the node and element data for input to the finite element code used. Using the finite element code, loads will be applied to produce a pure bending moment about the joint. The load will be incremented and springs will be eliminated as they go into tension. Joint rotations will be calculated and plotted against the applied moment to develop a moment-rotation curve.

A joint physical model will be built and tested to provide experimental results to compare with the numerical ones. The test setup will be like that shown in Figure 5-24. The experimental results will be used to verify the finite element model.

Once the technique of estimating joint stiffness using the finite element model has been validated, then a finite element model of this actual joint can be constructed to obtain the actual stiffness to be used in the ring model.

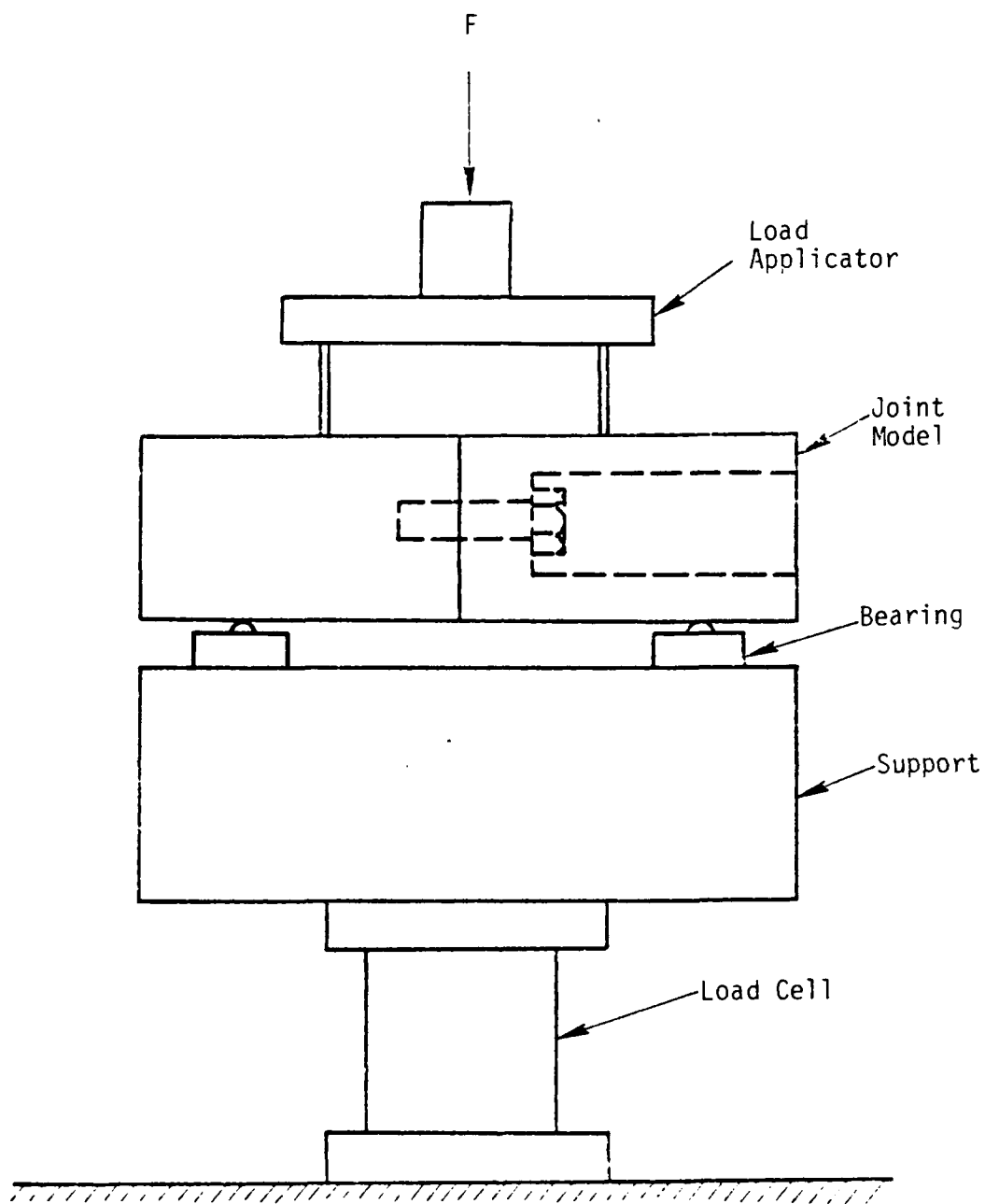


Figure 5-24. Joint Test Apparatus

CHAPTER 6

MICROASPERITY LUBRICATION

Background

Previous work [5] shows that at very low speeds (5-50 mm/sec) seal friction is very high. As the speed increases above this value, friction drops off very rapidly (specific results are shown later). This result is the same for both wavy and flat faced seals. It applies to the case of a carbon graphite sliding against a hard surface with mechanical contact in water.

This behavior is important to seal performance in two ways. First the high friction at low speed must be allowed for in the mechanical design of a seal because it represents the startup condition. Also some seals, such as in submarines, operate at a very low speed for a significant fraction of their life. Since this low speed friction can be as much as ten times the friction at higher speeds, such considerations must be made. The second reason for the importance of friction reduction with speed is that this reduction may in fact make it possible to operate seals of moderate and higher speeds. If the friction did not decrease, the friction power at higher speeds would become prohibitively large, and seals would heat check, wear, and score much quicker than is the actual case. Thus, this friction characteristic is very beneficial.

It was thought in the early part of this investigation that the friction reduction observed is due to hydrodynamic effects caused by

waviness. The flat face variable speed tests reported in the 1981 report [5] refutes this idea. In fact it is to be noted from those results that the experimental reduction in friction is much greater than that predicted by hydrodynamic effects caused by waviness.

After these observations were made it was decided that an investigation of speed caused friction behavior in carbon-hardface-water lubricated sliding systems should be pursued in some detail for the following reasons:

- 1) Good quantitative speed versus friction and wear data was needed for seal design purposes.
- 2) The cause of the friction reduction was not understood.
- 3) If the mechanics of the reduction could be understood, it might be possible to enhance it and reduce seal friction (and wear) even further.
- 4) If this behavior could be properly quantified, then it would make the task of assessing hydrodynamic effects in water much easier.

To fulfill these objectives, a study was undertaken to better understand sliding friction in carbon-hardface-water systems.

Theory

Approaching the phenomenon from a lubrication standpoint, if one considers theoretically flat parallel surfaces sliding parallel to each other and separated by an isothermal, uniform, steady film of Newtonian fluid, it can be shown using classical lubrication theory that no fluid films pressure is generated which might cause a friction reduction. On

the other hand, considering this phenomenon as being caused by a reduction of mechanical friction between the two materials as caused by the generally accepted adhesion theory of friction, then one must rationalize that adhesive bond shear strength must somehow decrease with increasing speed. No known theory suggests any strong relationship between adhesion bond strength and speed. However, it is widely known in dry sliding that starting friction is higher than moving friction. In the case of lubricated sliding this difference as reported in Reference [48] for many cases is not nearly as pronounced as has been measured in this present work.

Thus while it is possible that some of the observed behavior is caused by changes in the mechanical contact friction; the results show that friction coefficient approaches 0.01 with increasing speed. No known dry friction is this low, so that the reduction in friction cannot simply be a reduction of adhesive bond strength. Thus some type of hydrodynamic mechanism must account for all or a part of this reduction. This notion is supported by the fact that similar observations to the present have been made for parallel surface oil lubricated thrust bearings.

Many explanations have been advanced for the reduction in friction with speed in lubricated parallel sliding systems. Nearly all of the explanations hold that true parallel surfaces cannot be achieved in reality or that nonparallel surface geometries are generated by the process itself. Thus such load support can be explained by temperature induced warpage, unplanned machined waviness, eccentric rotation, seal

wobble and bounce, lubricant density changes, non-Newtonian lubricant effects, and microasperity lubrication.

Microasperity lubrication occurs when surface asperities act as small hydrodynamic thrust bearings. All real surfaces have asperities which can act as microscopic hydrobearings and provide lift. Because carbon has many natural asperities and because most of the other possible causes were minimized in the experiments, it was hypothesized that microasperity lubrication is the mechanism which causes the large friction reduction in the tests cited previously. The remainder of this chapter explains how this hypothesis was tested.

Microasperity Lubrication

The application of classical lubrication theory to a symmetric asperity like the one shown in Figure 6-1 yields a pressure distribution similar to the one depicted. Since the pressure distribution is an odd function, integration over the asperity to obtain load support yields zero. In actuality, the film pressure cannot become negative (fluids cannot support tensile stress) and the liquid cavitates when the pressure drops below the cavitation pressure. This situation is shown in Figure 6-2. Integration of the truncated pressure distribution yields a net positive load support.

Research reported by Hamilton et al. [49] clearly shows the relationship between microasperities and load support in parallel face rotary seals. An experimental seal with an optically transparent rotor was constructed so cavitation steamers could be observed in the oil lubricating film. In Hamilton's first test the torque was initially

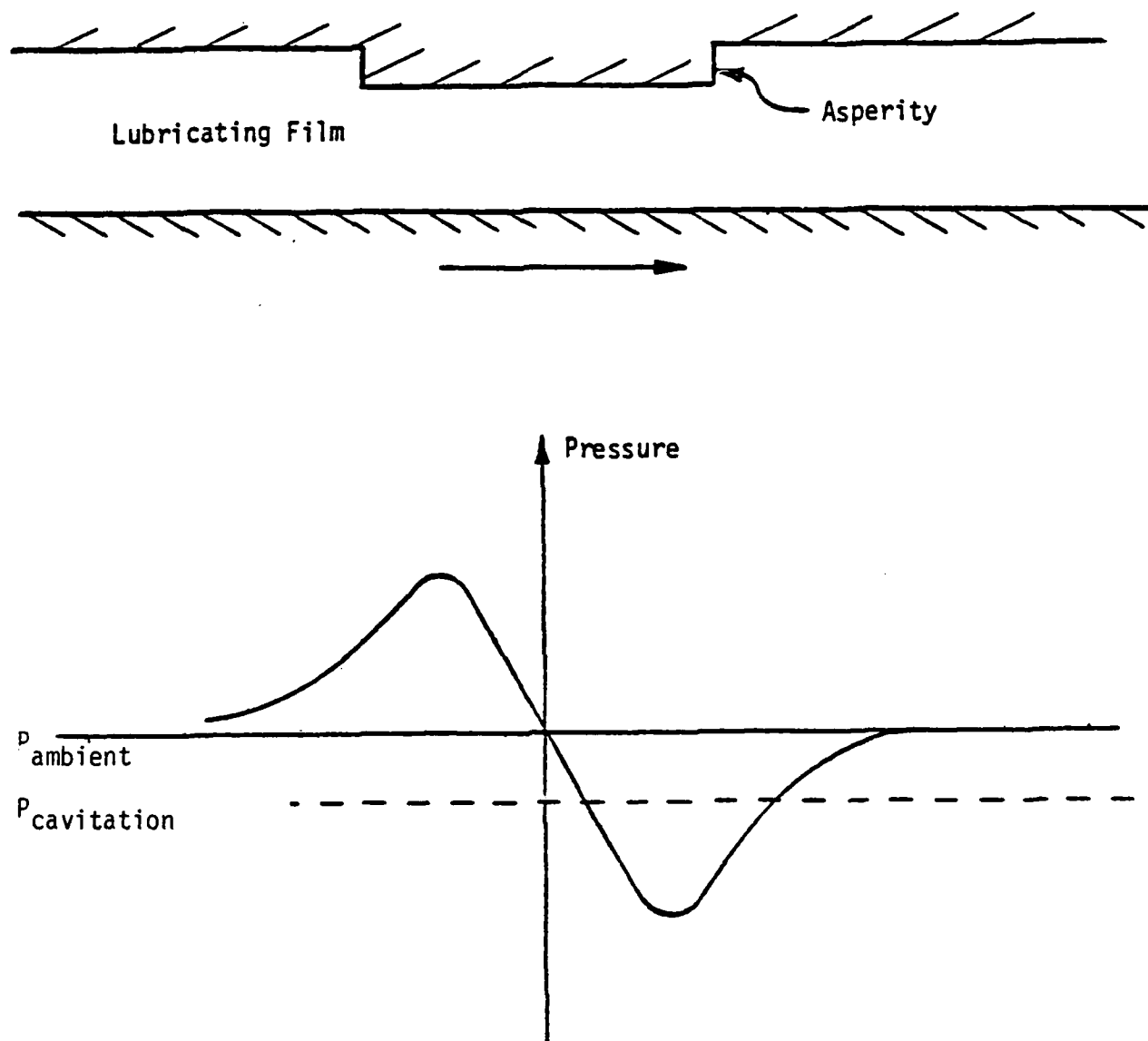


Figure 6-1. Asperity Pressure Distribution Predicted by Classical Lubrication Theory.

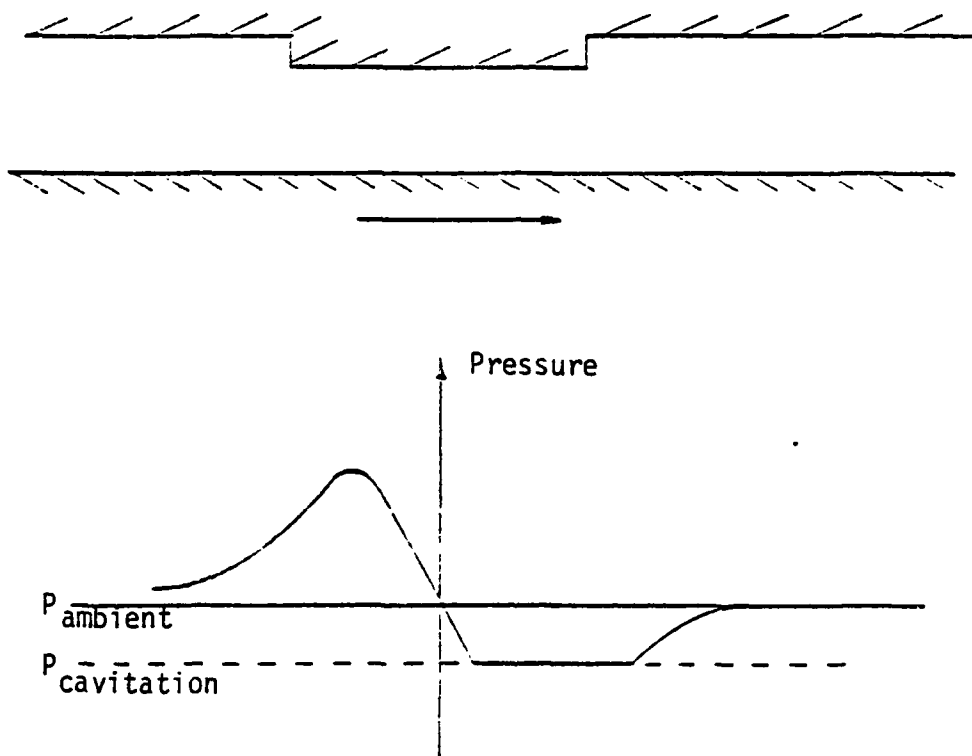


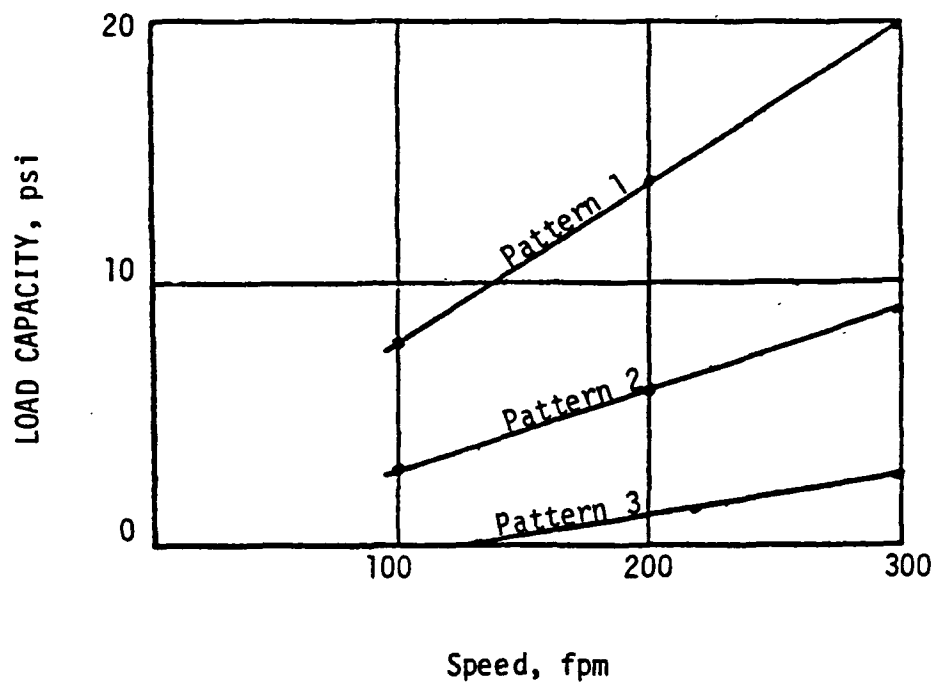
Figure 6-2. Truncated Asperity Pressure Distribution.

high and no cavitation streamers were observed. About one minute after start up the friction torque dropped sharply and cavitation streamers appeared simultaneously. Later it was found that the bands of cavitation occurred in areas where fine scratches or irregularities existed.

Next, asperity patterns were photoetched into the stator and tested using a hardened steel rotor. A known face load was applied and the film thickness measured by passing a very small alternating current of 7 kHz through the film. The face load was increased until a partial or complete destruction of the ac-film voltage drop occurred. At this point the film was considered to be nonexistent and load capacity reached. The test results indicated that there is a relationship between asperity distribution and geometry, speed, and load capacity. These results are shown in Figure 6-3.

Hamilton et al. [49] also presented a theoretical analysis and obtained an approximate solution for the load support generated by a field of asperities. The analysis begins with the Reynolds equation in polar form applied to the asperity shown in Figure 6-4. The result is then applied to the asperity field also shown in Figure 6-4 and then integrated to obtain load support. This model accounts for the existence of cavitation and shows load support to be a function of ambient pressure. This mathematical model is used as a comparison to the experimental results in this work.

The derivation of Hamilton's [49] model begins with the Reynolds equation in polar form. The major assumptions made are:



Experimental load capacity. Lubricant: 13.5-centistoke (100°F) mineral oil; stator-surface temperature, 100°F; inside sealed pressure, 5 psig.

Pattern	Asperity diameter, mils	Number of Asperities per sq. in.	Fraction of Area Covered by Asperities	Average Asperity Height, Microin.
1	4.6	12100	0.21	100
2	1.2	22500	0.026	83
3	1.7	22500	0.051	130

Figure 6-3. Experimental Results from Hamilton et al [9].

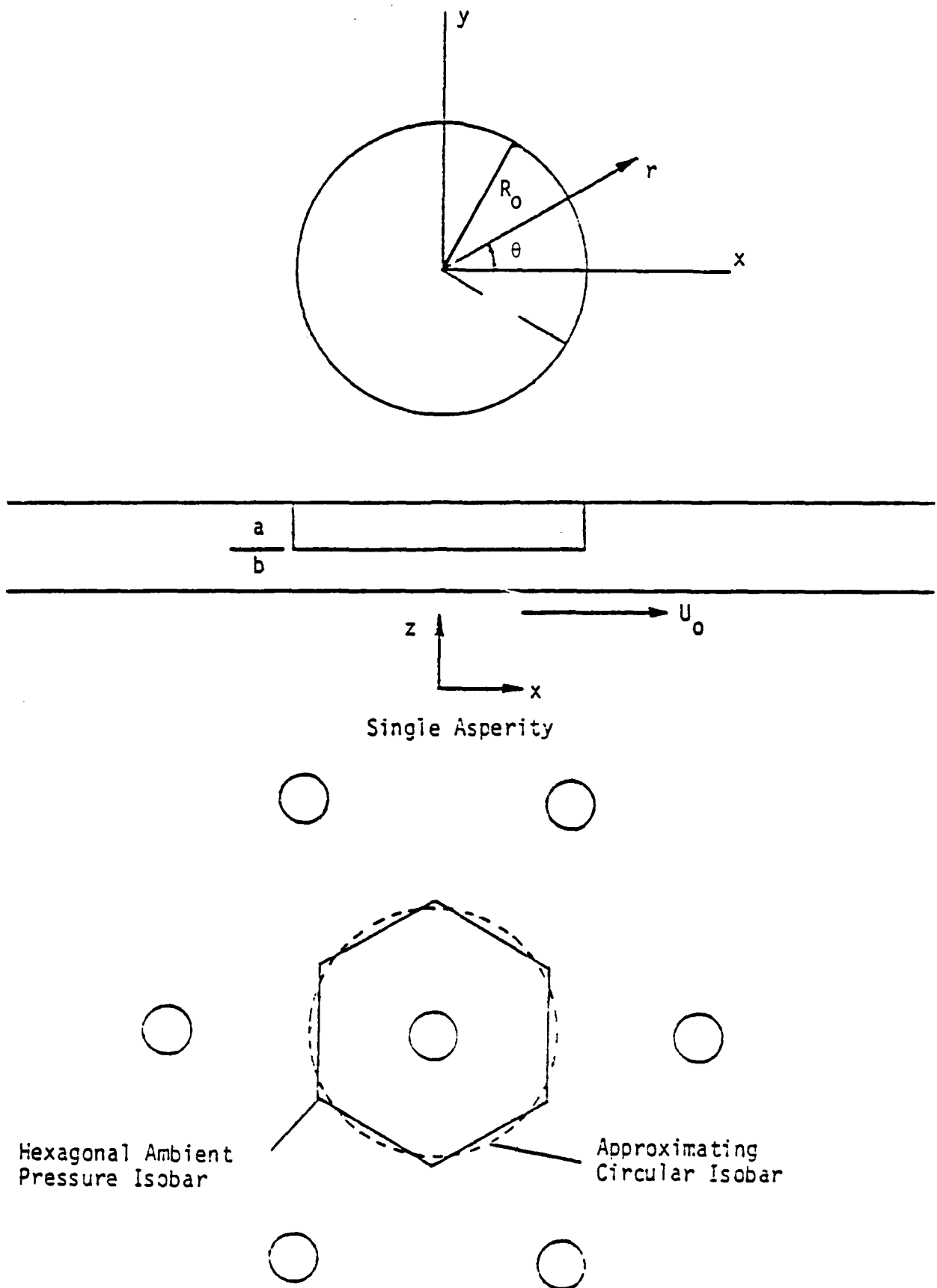


Figure 6-4. Asperity Geometry from Hamilton et al [49].

- 1) The Reynolds equation and all the assumptions made in its derivation apply.
- 2) Pressures calculated below the cavitation pressure of the lubricant are replaced by the cavitation pressure.
- 3) Load support is obtained by integration of the resulting truncated pressure distribution.

With reference to Figure 6-4

P = lubricant pressure

P_{∞} = ambient pressure

P_c = cavitation pressure

p = amplitude of $P-P$

q_{θ} = tangential volumetric flow rate

q_r = radial volumetric flow rate

μ = absolute viscosity

δ^2 = area fraction occupied by asperities

W = load support

$h, a, b, R_o, u_o, r, \theta$ as defined by Figure 5

From Figure 6-4

$$h(r) = \begin{cases} b, & r < R_o, \quad \text{Region I} \\ a + b, & r > R_o, \quad \text{Region II} \end{cases} \quad (6-2)$$

The final result of Hamilton's work is

$$w = \frac{8 \mu V_o R_o a \delta}{\pi (b^3 + \gamma (a + b)^3) (1 + \delta)} - \frac{1}{2} (P_{\infty} - P_c) . \quad (6-3)$$

This equation shows that load support can be readily developed by micro-asperities.

Other investigators have considered microasperity lubrication. Kojabashian and Richardson [50] investigated the surface topography of the mating surfaces in carbon face seals. The experimental results reveal the existence of a micropad structure that develops with wear. This structure consists of flat raised pads highly variable in size. It is shown that these pads can be adequately described by an exponential probability distribution. These micropads are modeled by an equivalent step bearing and used to predict the performance of carbon seals. It was found that using average pad sizes in the analytical model causes hydrodynamic load support to be underestimated by an order of magnitude. It is concluded that more work is necessary before the micropad model can be definitely established or refuted.

Kistler et al. [51] have concentrated on the cavitation phenomenon in face seals. The objective of this work was to experimentally and theoretically study the effects of cavitation in thin lubricating films between rough surfaces. Presented in this work is the concept of a pressure distribution being truncated by cavitation. A model presented by Patir and Cheng [52] that considers the effects of surface roughness is modified to include the effects of cavitation. This model used to generate a pressure distribution between two rough surfaces sliding parallel to each other. The resulting pressure distribution is used to generate a plot of load capacity versus ambient pressure and clearly shows that load capacity is developed by an asperity field.

Test of Hypothesis

Microasperity lubrication has never been investigated in water sliding systems. While in oil systems the work of Hamilton [49] provides a virtual certainty that microasperity effects do exist and provide load support, no such verification exists for water. Of course, previous equations and work both indicate that microasperity lubrication should occur, although its effect will be smaller than for oil because of the reduced viscosity.

To test the existence of microasperity lubrication in a real system is difficult. (Note Hamilton contrived microasperities.) However an unusual approach was found. If the hypothesis that the fluid pressure is an odd function and is truncated by cavitation is true in microasperity lubrication, then ambient pressure should have an effect on load support. For example, if the ambient pressure is raised, less of the pressure distribution will fall below the cavitation pressure, resulting in a lower load support. If the ambient pressure is raised to a high enough level such that the asperity fluid pressure never drops below the cavitation pressure, the load support will be zero (see Figure 6-5). Thus, changing the ambient pressure and observing the resulting load support can be used to experimentally investigate the effects of microasperities on hydrodynamic lubrication. The load support need not be directly measured; instead, a constant load is applied and friction torque is measured. An increase in friction is interpreted as a decrease in hydrodynamic load support.

This approach is supported by Hamilton's equation. Figure 6-6 shows predicted load support as a function of ambient pressure at

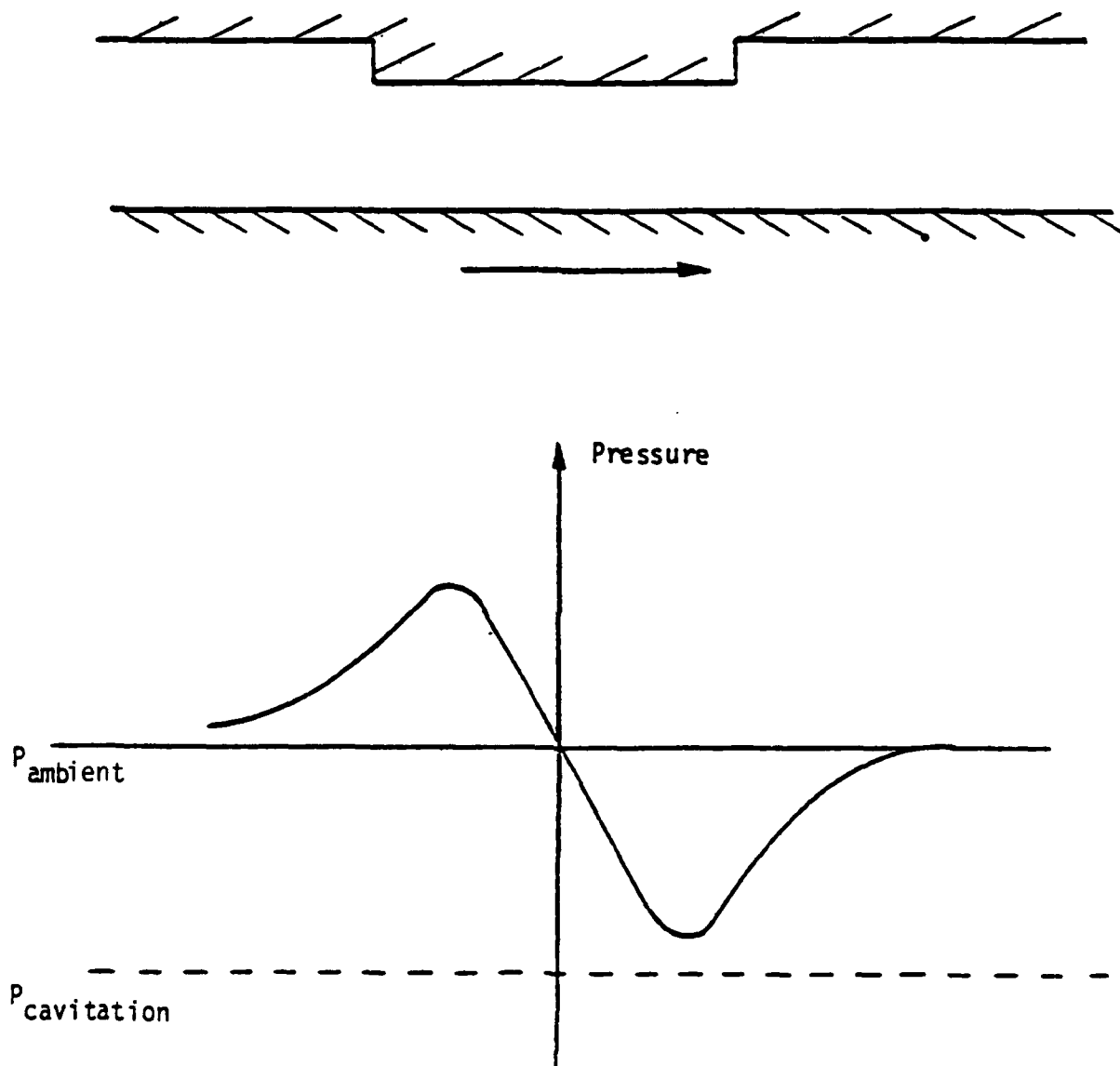


Figure 6-5. Zero Load Support.

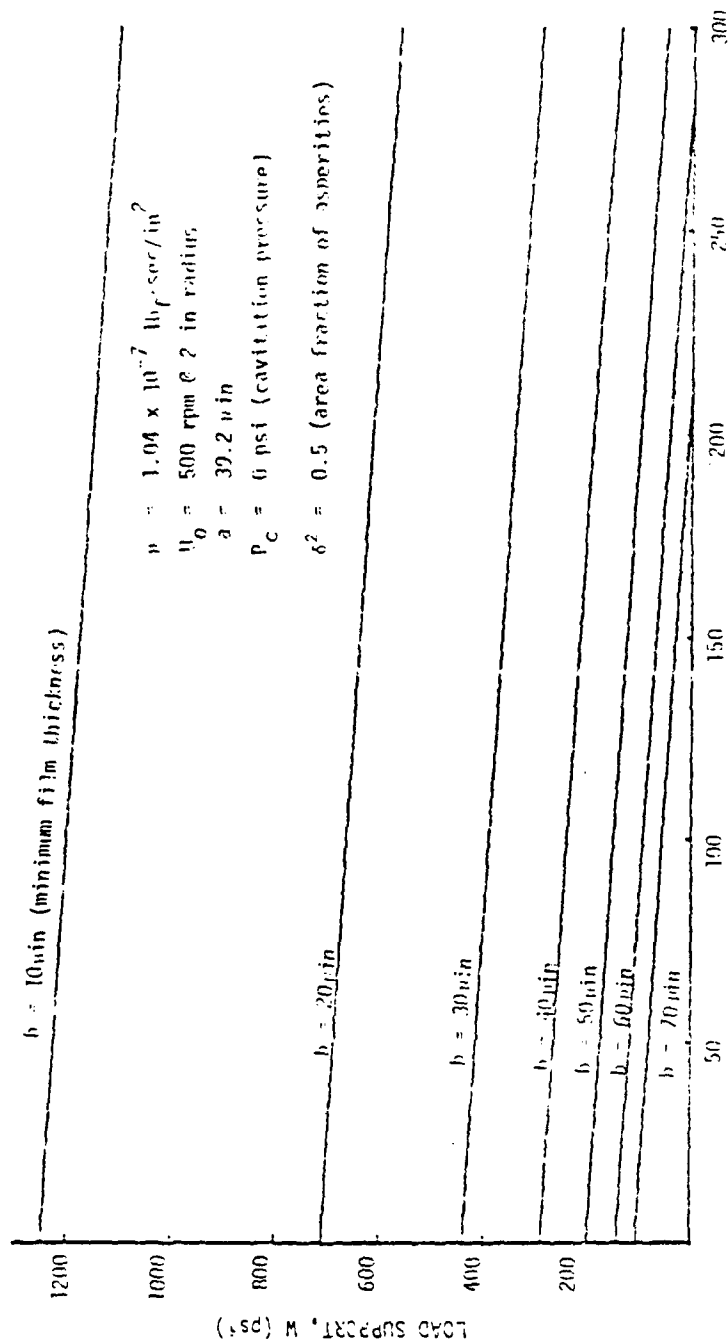


Figure 6-6. Load Support vs Ambient Pressure (from Hamilton) [49].

constant minimum film thickness. As ambient pressure is increased the decreased load support would require that more of the load be supported by mechanical contact and friction would therefore increase. Kistler [51] shows a similar result in Figure 6-7. Thus, experimentally operating a parallel sliding experiment in the presence of a variable ambient pressure should provide an indication of whether or not micro-asperity lubrication is functioning.

Experimental Apparatus

To test the stated hypothesis, an apparatus was designed to measure friction under the following conditions:

- 1) Parallel sliding in water.
- 2) Carbon versus hard face material.
- 3) Tilt of the face and wobble of the rotor to be minimized.
- 4) Water can be pressurized to 300 psi.
- 5) Variable speed from 0.5 to 500 in./s.
- 6) No hydrostatic load support component (unlike in face seals).

An experimental apparatus designed by Summers and Lebeck in 1981 [5] for the purpose of measuring dynamic coefficients of friction of carbon materials rubbing on hard faced materials in water under pressure was used to meet the basic condition and criteria above. This apparatus consists of a rotating support system, stationary support system, pneumatic load unit, strain gage unit, and a pressure vessel (see Figure 6-8).

The rotating support system is driven by a constant torque variable speed DC motor and the rotor (rubbing surface) is a tungsten

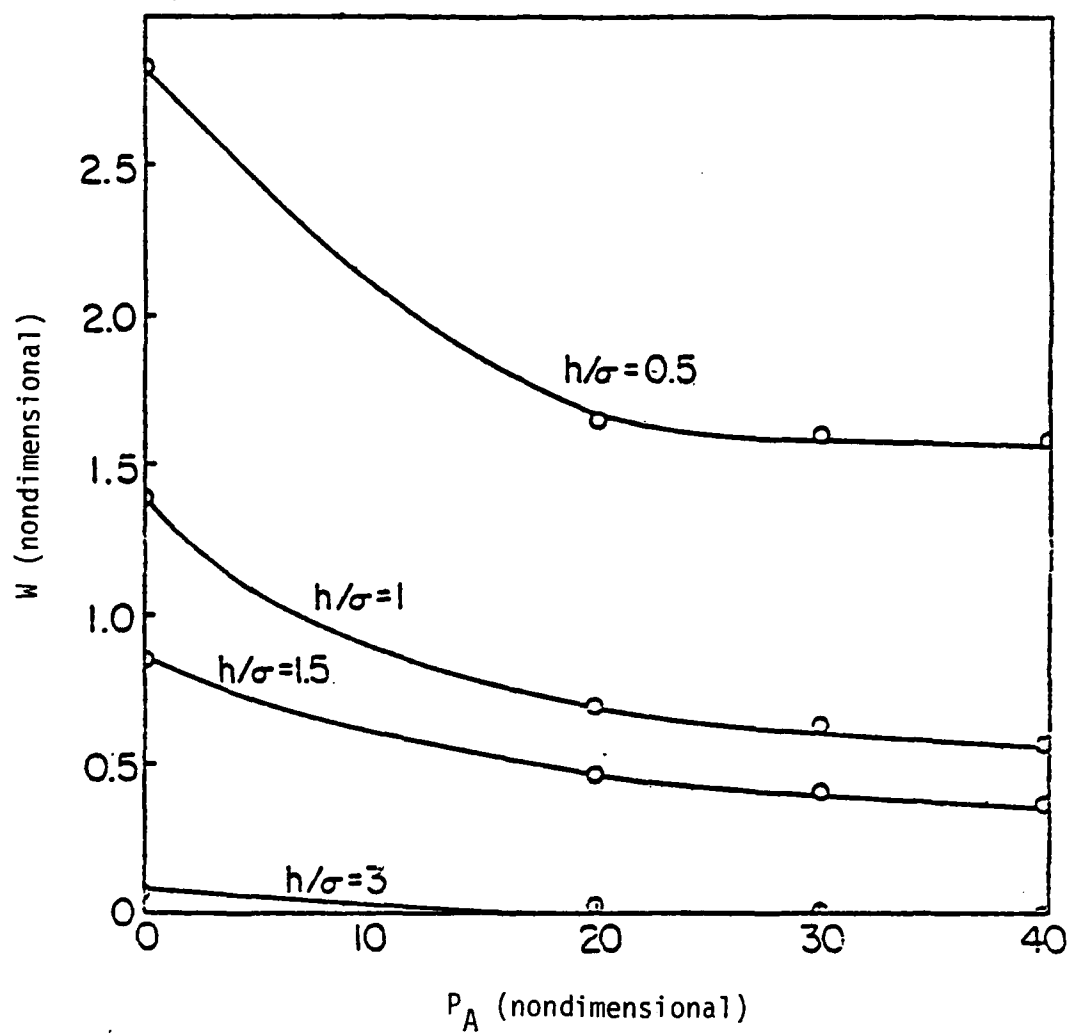


Figure 6-7. Load Support vs Ambient Pressure (from Kistler) [3].



Figure 6-8. Friction and Wear Test Apparatus.

carbide seal ring. The axial runout of the rotor at the rubbing radius was reduced to 0.0002 in. total indicated reading to minimize hydrodynamic load support generated by wobble and bounce. This was accomplished by grinding the rotor supporting mount in the bearings that are used in the tester itself.

The stationary support system holds three carbon dowels on the rotor face. The faces of the carbon dowels are always held parallel to the face of the rotor even after wear occurs, thus eliminating hydrodynamic load support that would be generated if these surfaces were not parallel. This geometry is maintained by supporting the stator (the carbon dowels) with a flexible rod which can deflect to compensate for uneven wear and misalignment. A combination of the three point contact geometry of the carbon dowels and the flexible rod assures that the plane formed by the carbon dowel rubbing surfaces will be parallel to the plane of the rotor rubbing surface when a load is applied to the flexible rod.

The torque measuring and load application parts of this original apparatus were completely redesigned herein because it was discovered early in this investigation that the O-ring over the flexible rod caused indeterminate friction both for the torque measurement and for load application.

Initially two designs were considered. The first reduces O-ring size and separates the stator from the rod through which the load is applied, the objective being to reduce friction at the O-ring seal and make the measured friction torque independent of the vessel pressure (see Figure 6-9). The contact radius between the load rod and the

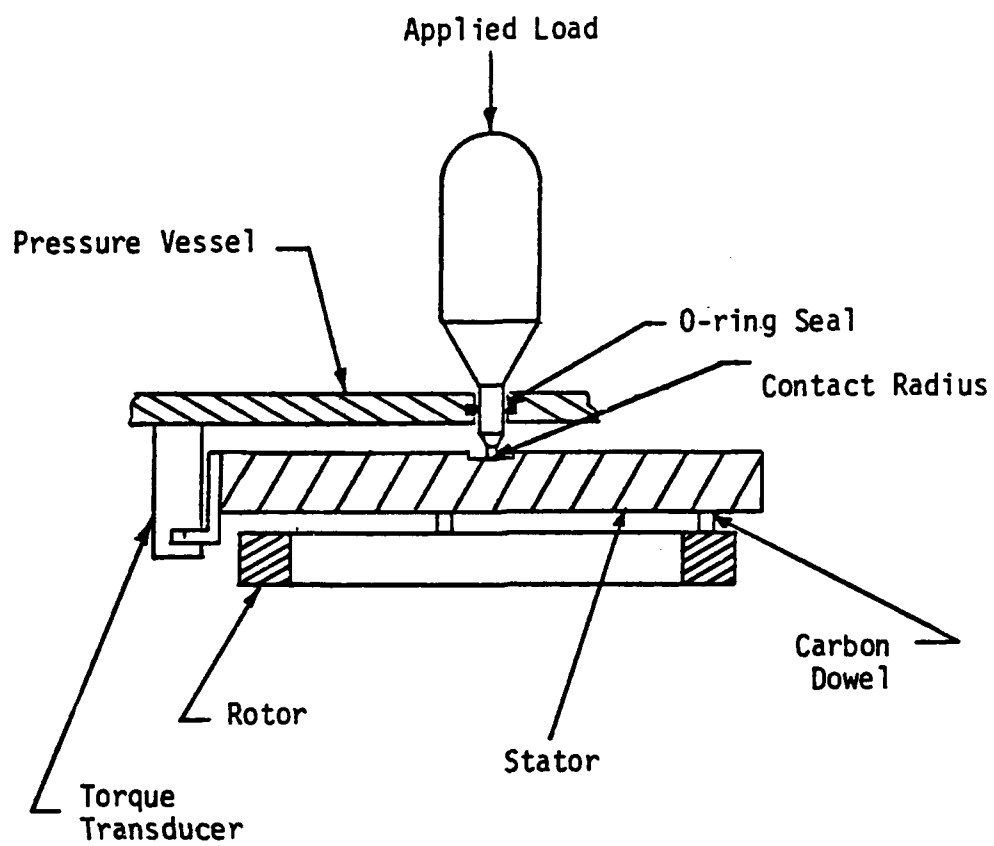


Figure 6-9. Point Load Design

stator is minimized in order to reduce the friction torque generated there. This contact radius is limited by the stress developed in the small diameter rod used to make this contact. Since a small contact radius is desired to minimize the unwanted friction, the stress developed in the small diameter used to make this contact is extremely high. This design would not make the measured friction torque independent of vessel pressure. The friction developed at the O-ring is reduced but still a function of vessel pressure in an unknown way making the applied load an unknown function of vessel pressure. This design would make it necessary to measure friction torque inside the pressure vessel as shown in Figure 6-9.

The second design considered incorporates a hydrostatic bearing in place of the flexible rod and creates a zero friction liquid coupling between the applied load and the stator (see Figure 6-10). The hydrostatic bearing assures that a fluid gap exists between the bearing parts. Since there is zero speed between these parts, there is zero friction. This design requires the torque transducer to be inside the pressure vessel. This configuration makes measured friction torque independent of vessel pressure but the applied load is still made through an O-ring seal and therefore dependent on vessel pressure in an unknown way. However this problem can be avoided by including a means for measuring the applied load inside the pressure vessel. It was decided to use the hydrostatic bearing design because this design totally eliminates torque error caused by unwanted friction.

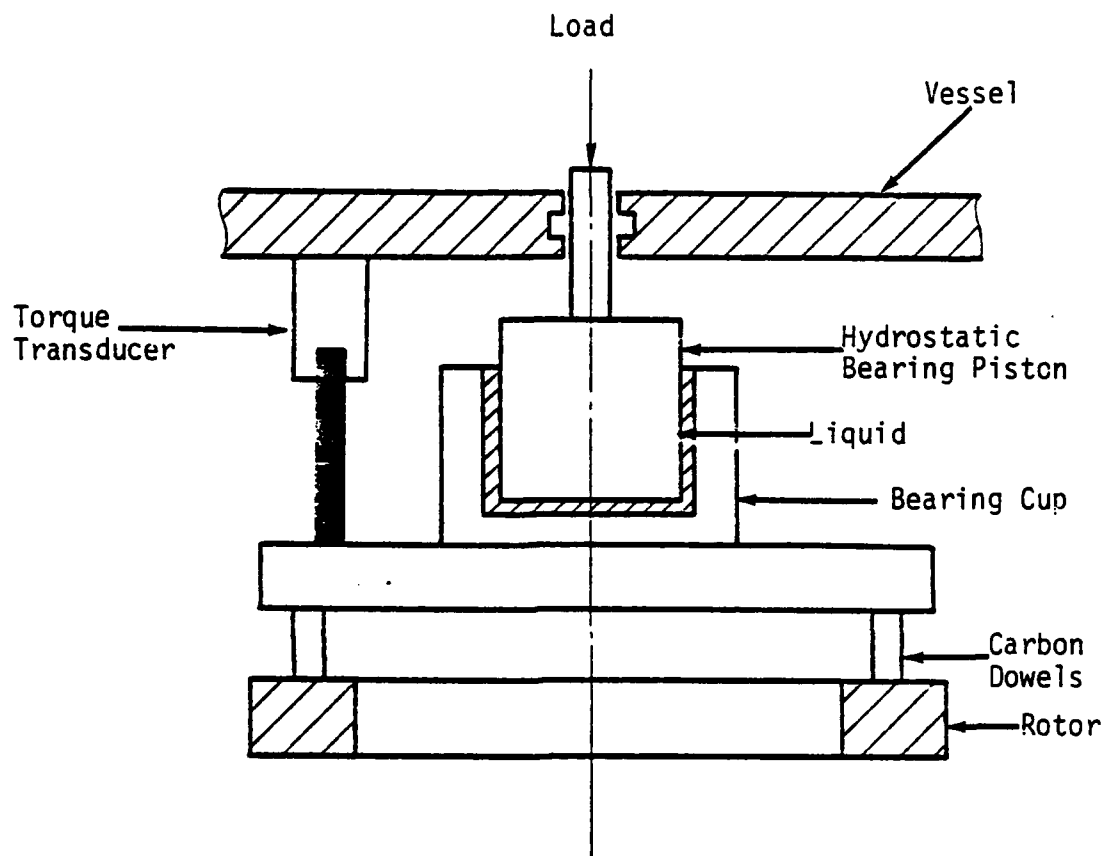


Figure 6-10. Hydrostatic Bearing Support.

Hydrostatic Bearing Design

The hydrostatic bearing designed must be capable of supporting normal and radial loads. The normal load is the applied load. The radial load comes from the torque reaction and unequal friction on the three dowels. A piston shaped bearing with hydrostatic pads on the sides and bottom (see Figure 6-11) is used. The piston fits into a cylindrical bearing cup. This particular bearing design has two sets of four pads each spaced axially some distance apart. The purpose of the four pads is to provide radial stiffness in all directions as shown in Figure 6-12. The two sets are used to provide angular stiffness (Figure 6-12). The axial load is carried by the one pad at the bottom.

The details of the design procedure are contained in Reference [29]. Many factors were considered:

- 1) Radial stiffness, angular stiffness, axial stiffness
- 2) Total flow
- 3) Minimum clearance
- 4) Deformation of the bearing parts
- 5) Effects of misalignment

Capillary compensation was chosen. Each bearing on the side is stiff by itself but always acts opposite to another so that it tends to center the piston. The operating clearance was selected at 0.00075 in. (0.02 mm) to minimize flow yet be large enough to get good separation.

The two essential parts of the design are shown in Figure 6-13 and 6-14. An assembly is shown in Figure 6-17. Capillary tubes are shown in Figure 6-11. In order to assure that the stator dowels are held

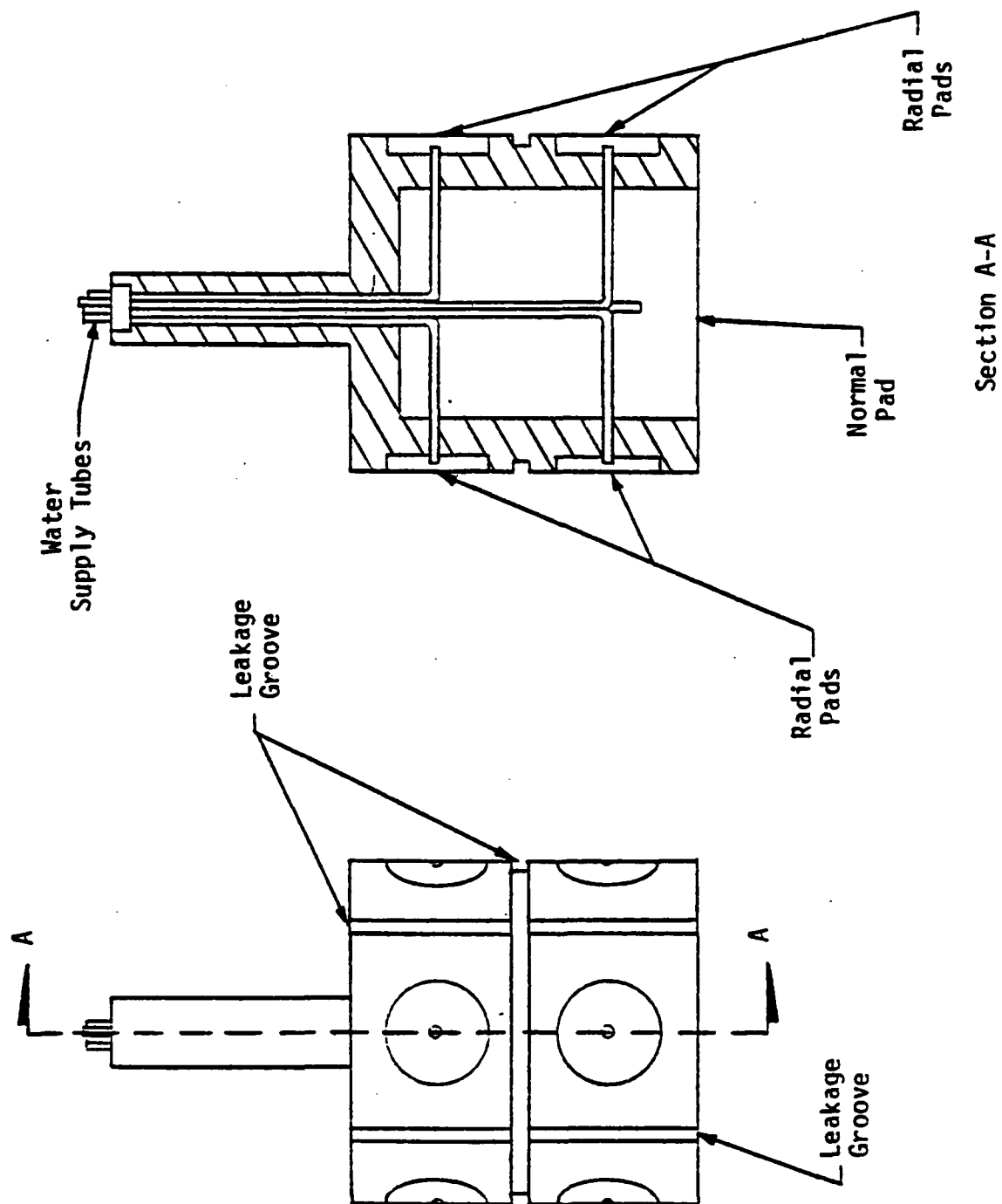
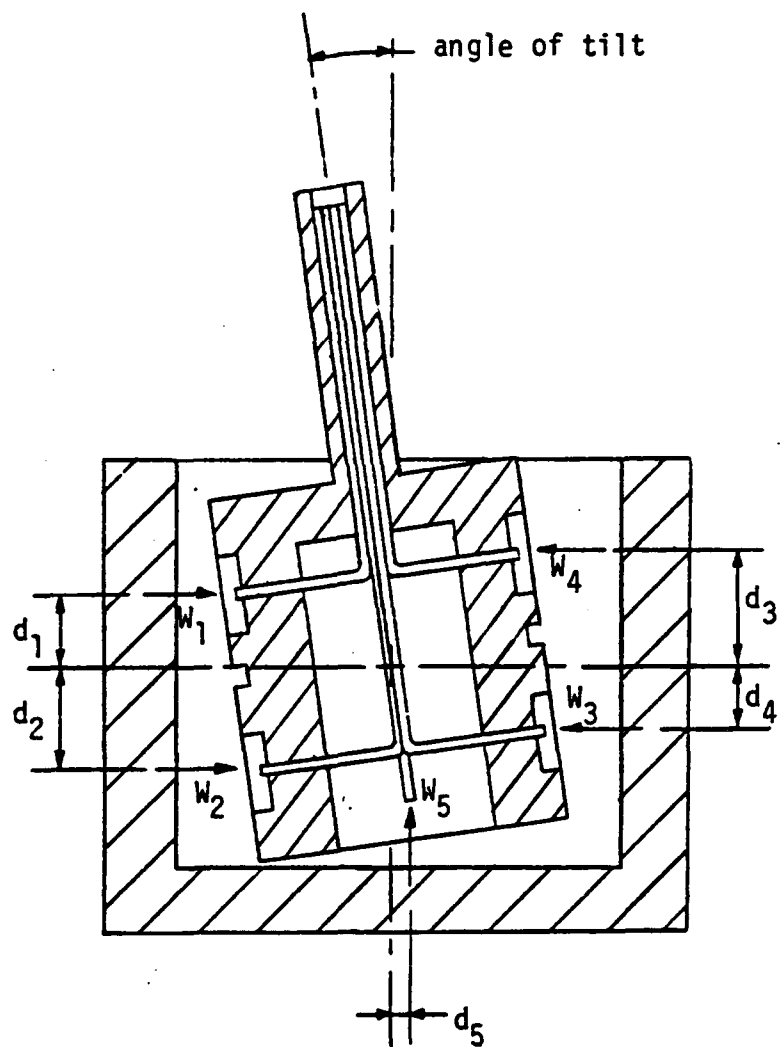
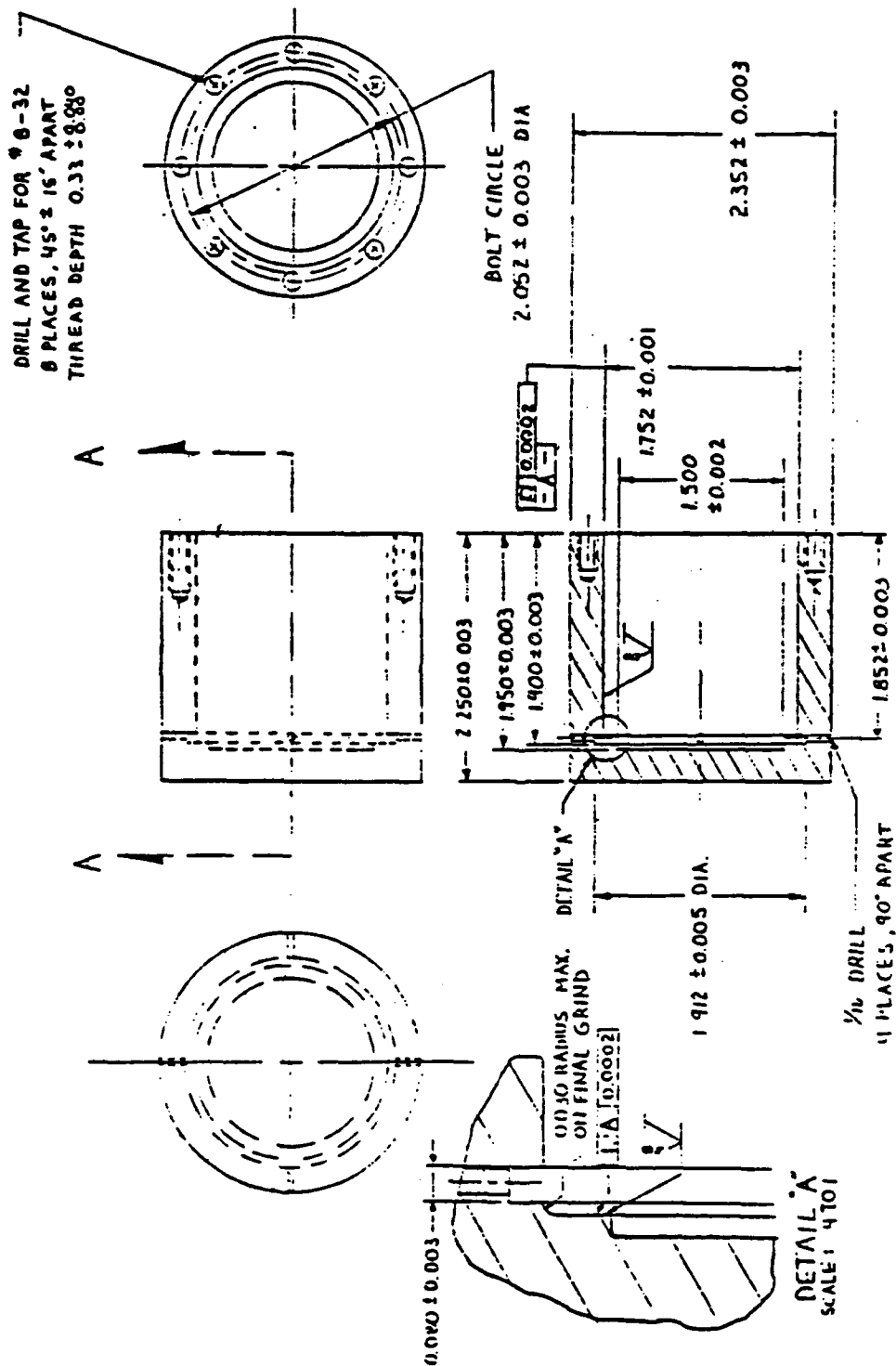


Figure 6-11. Hydrostatic Bearing Piston.



$$\text{Restoring Moment} = W_1 d_1 + W_3 d_3 - W_2 d_2 - W_4 d_4 - W_5 d_5$$

Figure 6-12. Bearing Misalignment and Moment Generated.



NOTES:

- 1) BREAK ALL SHARP CORNERS
- 2) ALL DIMENSIONS IN INCHES UNLESS OTHERWISE SPECIFIED

SECTION A-A

- 3) MATERIAL: 420 STAINLESS HEAT TREATED TO OBTAIN 200 KSI YIELD STRENGTH

Figure 6-14. Hydrostatic Bearing Cup.

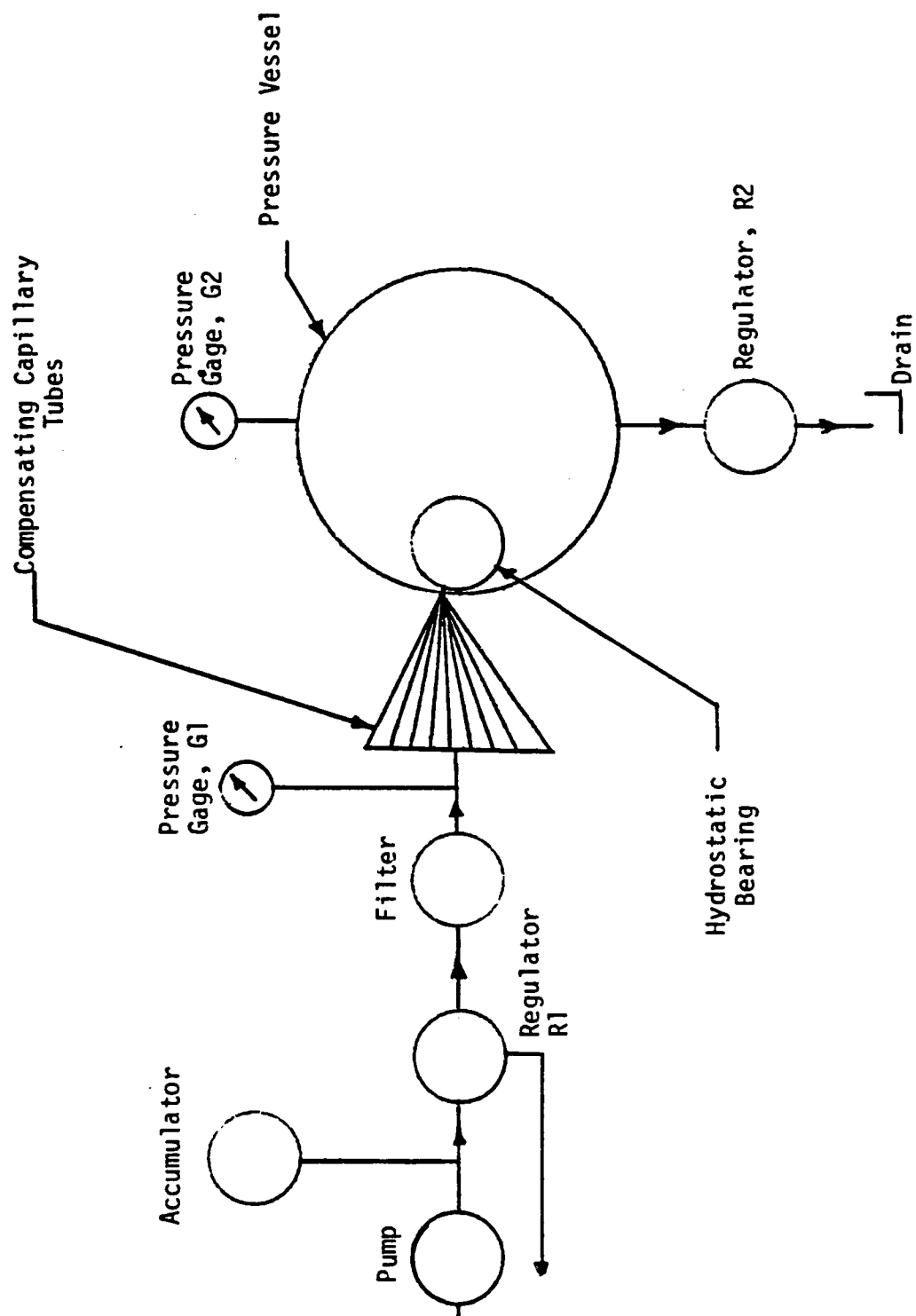


Figure 6-17. Water System Plumbing Diagram.

flat on the rotor, regardless of shaft misalignments through the pressure vessel, it is necessary to couple the hydrostatic bearing with a spring of much smaller stiffness. Figure 6-16 illustrates this configuration schematically. The flexible coupling is accomplished using an annular diaphragm shown in Figure 6-15. This diaphragm must be less stiff than the hydrostatic bearing and resistant to a water environment.

The friction torque reaction generated at the rotor-stator interface is transmitted directly from the stator to a cantilever beam mounted on the pressure vessel bulkhead (see Figure 6-15). This cantilever beam has a strain gage bridge mounted on it to measure bending strain. The cantilever beam is designed to have as much bending strain as possible to maximize the torque transducer output and increase torque resolution.

The piston must pass through the pressure vessel bulkhead where it is externally loaded with a pneumatic load unit. Therefore the piston must pass through an O-ring where vessel pressure dependent axial friction is generated. This is undesirable because it makes the load applied to the stator dependent on vessel pressure in an unknown way. This problem is minimized by reducing this O-ring friction to a minimum without ruining the integrity of the seal. This problem was later completely eliminated by measuring the load independently of the air pressure by placing strain gages on the diaphragm.

To insure that the bearing was working properly, the bearing cup was insulated electrically from the carbon holder ring and a wire was connected to the cup through the vessel. Then, when the cup was not

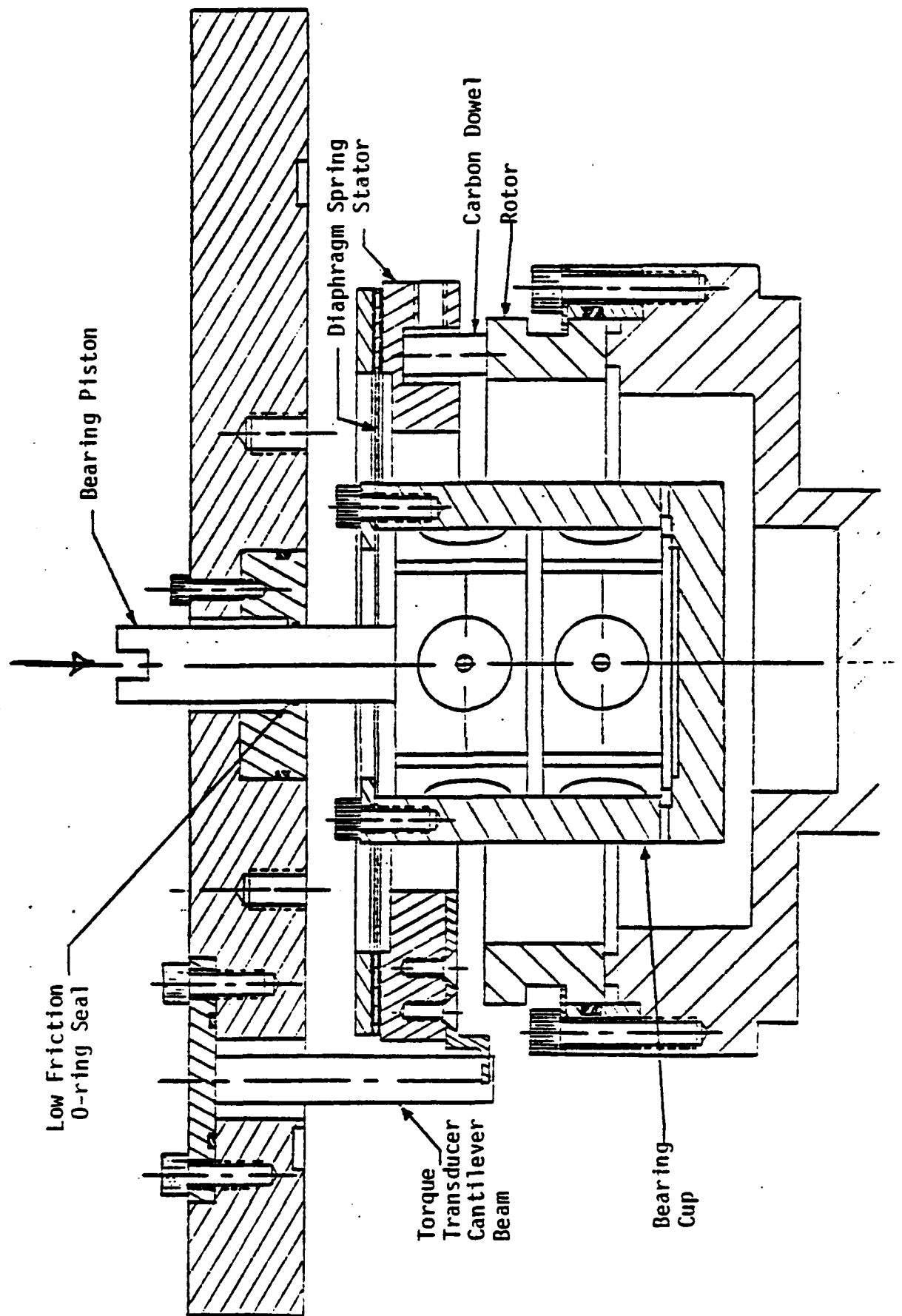


Figure 6-15. Hydrostatic Bearing Assembly.

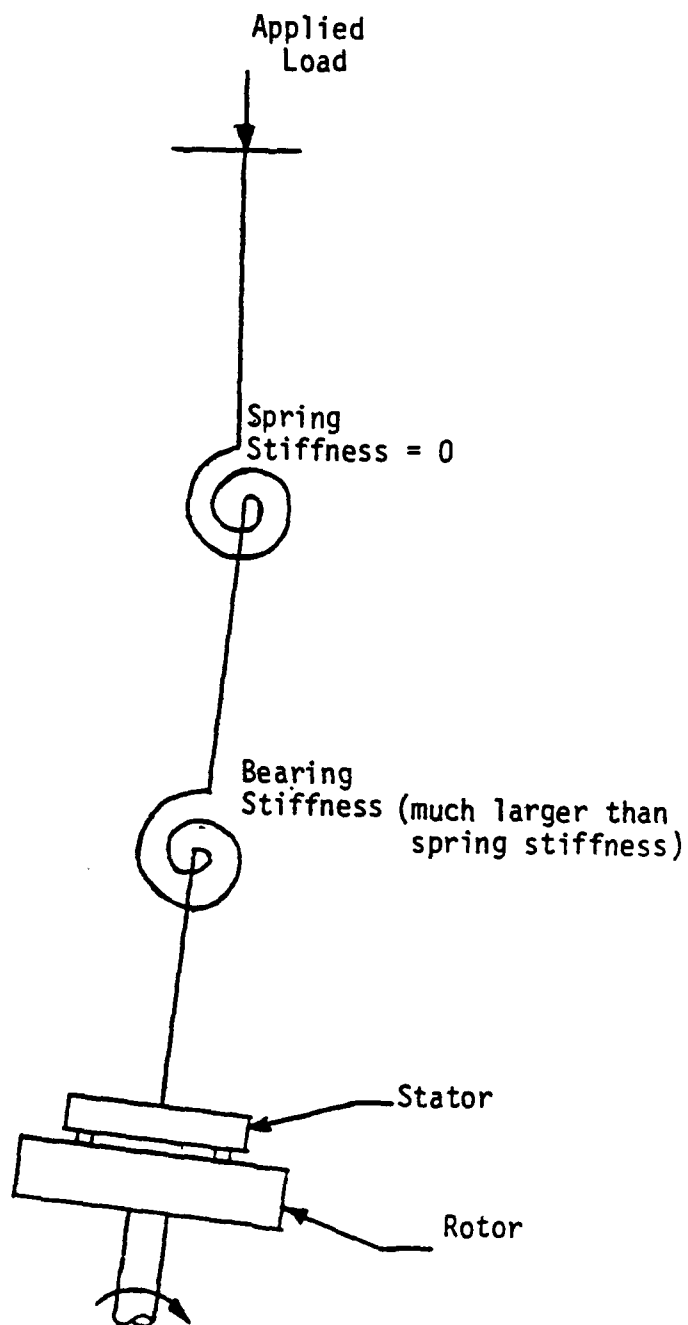


Figure 6-16. Torsional Spring Model.

touching, electrical resistance between the cup and the machine was large. When resistance became essentially zero, touching occurred.

Fabrication of the hydrostatic bearing was difficult because of the heat treating and precision required. Details of this process are contained in Reference [29].

Support System

A flow system was designed as shown in Figure 6-17. This arrangement allows the pressure drop through the capillary tubes and the bearing to be set at the desired level while permitting the vessel pressure (ambient) to be controlled. The need for the filter arises because clearances in the bearing are small (less than 0.00075 in.) and particulates can become jammed in the bearing causing mechanical contact to occur. The pump is a positive displacement triplex pump and delivers about 4 gpm at pressures up to 1200 psi. Pressure surges from the pump are controlled by the accumulator. The pressure on the inlet side of the bearing is controlled by back pressure regulator R1. Water is filtered to 2 μ m before entering the bearing. The vessel pressure is controlled by back pressure regulator R2. The pressure drop across the bearing and capillary tube is now the difference between the capillary tubes inlet pressure and the vessel pressure (G1-G2). To minimize the burden on the water filter, distilled water was used and recirculated as shown. Figure 6-18 shows the test apparatus assembly. The support equipment is behind the panel.

Data is collected with HP 9835B computer with internal printer and real time clock, an HP 9872A plotter, and an HP 6940 multiprogrammer

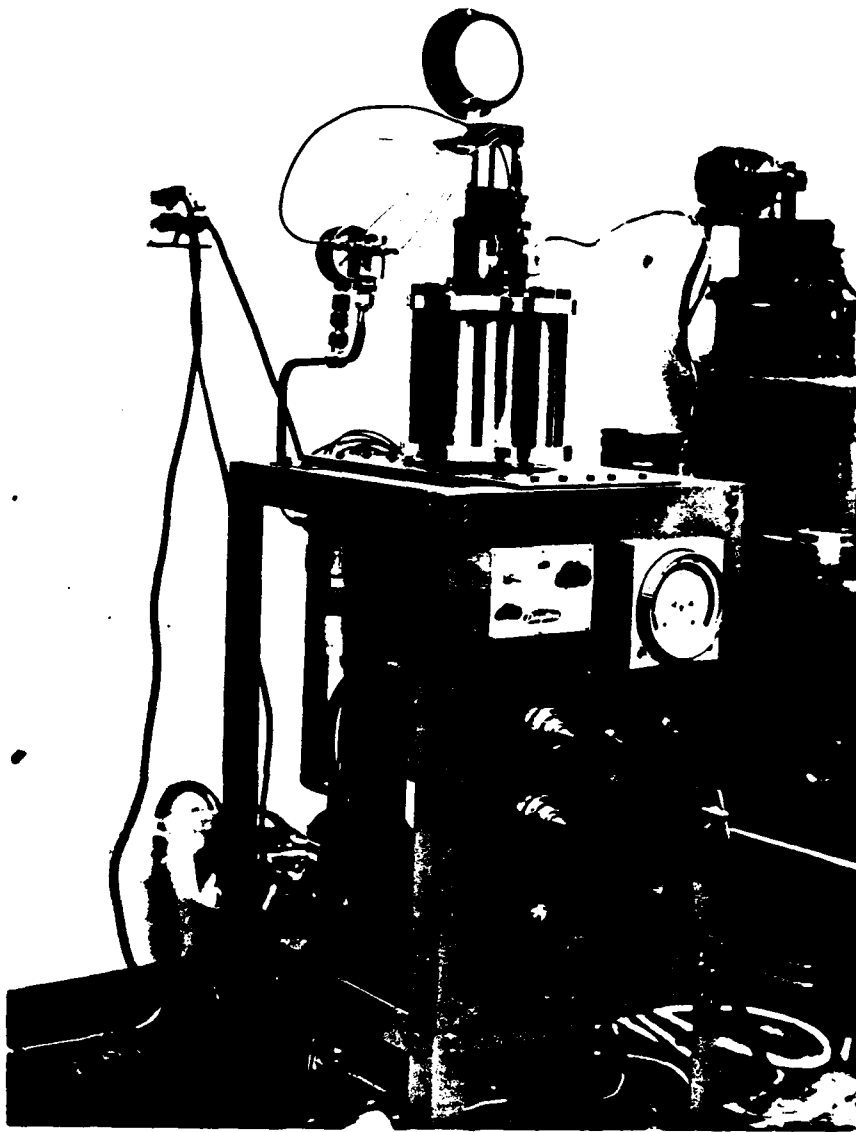


Figure 6-18. Friction Test Support System.

with a low level A/D card. Rotor speed, vessel pressure, flow rate, and applied load are all set manually.

Hydrostatic Bearing Tests and Calibration

The purpose of using a hydrostatic bearing in the friction and wear test apparatus is to create a frictionless fluid coupling between the applied load and the stator. Any friction generated in this interface will influence the friction torque being measured and invalidate the measurement, defeating the purpose of using a hydrostatic bearing. It is therefore necessary to monitor the bearing to be sure touching does not occur. As described, checking for bearing touching is accomplished by measuring the electrical resistance across the lubricating film in the bearing. If touching occurs the resistance will drop to zero. The results of the resistance measurement varied a lot over a period of time. The cause for this fluctuation was determined to be the result of a voltage generated across the film by ions in conjunction with using a DVM. Measurement showed this voltage to be about 1 μ V. Since the voltage generated by the digital meter to measure resistance is also about 1 μ V, the meter was being fooled by the voltage generated by the migrating ions and the resistance measurement is not correct.

It was discovered when operating the bearing that an analog resistance meter did not have this resistance fluctuation problem because this meter uses a much larger voltage in the measurement of resistance. The analog meter was therefore used for monitoring touching and the technique was considered reliable.

The first performance test consisted of applying a known load to the bearing with a 150 psi pressure differential across the capillary tubes and the bearing. The load was increased until touching occurred.

These tests showed that bearing operation was highly dependent on the cleanliness of the water and the bearing surfaces and that the load support of the bearing, indicated by the load at which touching first occurred, was less than predicted. This low load support prompted flow rate experiments to see if the flow rates predicted by the hydrostatic bearing model occurred in fact. It was determined that some redesign of the capillaries was needed [29]. After modifications were made, the bearing supported up to 170 lb normal load without touching. The model predicts that touching will occur with a 300 lb load. Recognizing that the bearing geometry was not perfect, this was considered adequate to start the test.

Calibration was accomplished by hanging weights with known values on the torque transducer cantilever beam and sampling 50 points with the data acquisition system described previously. An average of these points is then taken and a calibration constant (in.-lb/V) calculated. The calibration constant was 1.2 in.-lb/mV giving a least bit resolution of 0.006 in.-lb.

Testing of the force transducer on the diaphragm was performed by assembling the friction and wear test apparatus and applying an incrementally increasing load to the diaphragm. The resulting strain, read from the strain indicator attached to the bridge, was recorded for each load. Once the load reached 170 lb, the load was incrementally decreased and strains recorded. These tests showed the force

transducer's sensitivity to applied load is about 0.177 lb/microstrain. The force transducers sensitivity to tangential strain was shown to be insignificant. When the load was decreased from 170 lb, the strain readings were different from when the load was being increased by as much as 100 microstrain (18 lb). This was caused by the O-ring friction, and this demonstrates why the independent load measuring technique were necessary.

Experimental Procedure

The experimental procedure used to operate the modified friction and wear test apparatus is:

- 1) The capillary inlet pressure is set 300 psi above the desired vessel pressure (300 to 600 psi) using regulator R1 (see Figure 30).
- 2) Vessel pressure is set (10 to 300 psi) using regulator R2.
- 3) The torque transducer zero is obtained by reading the output of this transducer with no load applied.
- 4) The force transducer is zeroed by lifting the stator off the rotor and zeroing the strain indicator. The stator is separated from the rotor by pulling up on the piston.
- 5) If the test is the first in a series of tests with constant carbon pressure, the load is applied and the resulting strain on the diaphragm is recorded. For all other tests at this carbon pressure the load is applied by increasing the air pressure on the pneumatic load unit until the same strain

reading as on the first test at this carbon pressure is achieved.

- 6) When a test is finished the zero of the force transducer is checked to assure that no significant zero shift has occurred.
- 7) The desired rotor speed is set (50 to 2500 rpm) using a phototachometer.
- 8) The load applied to the stator (0 to 1000 psi carbon pressure) is adjusted using the air regulator on the pneumatic load unit.
- 9) Rotor speed is rechecked and readjusted if necessary.
- 10) The bearing film is monitored for a nonzero film resistance to assure that touching does not occur.
- 11) The computer program that runs the data acquisition system is run sampling 150 points/minute. Five minute tests are run.
- 12) Temperature control was achieved by circulating tap water of a constant temperature through a heat exchanger located in the distilled water reservoir.

Test Results

Using the tester as described before the hydrostatic bearing was added, many tests were run. Figure 6-19 shows a typical curve generated for friction as a function of speed. This curve is of questionable accuracy for the reasons mentioned but is included here to illustrate the trend of interest. When lower speeds were attempted friction

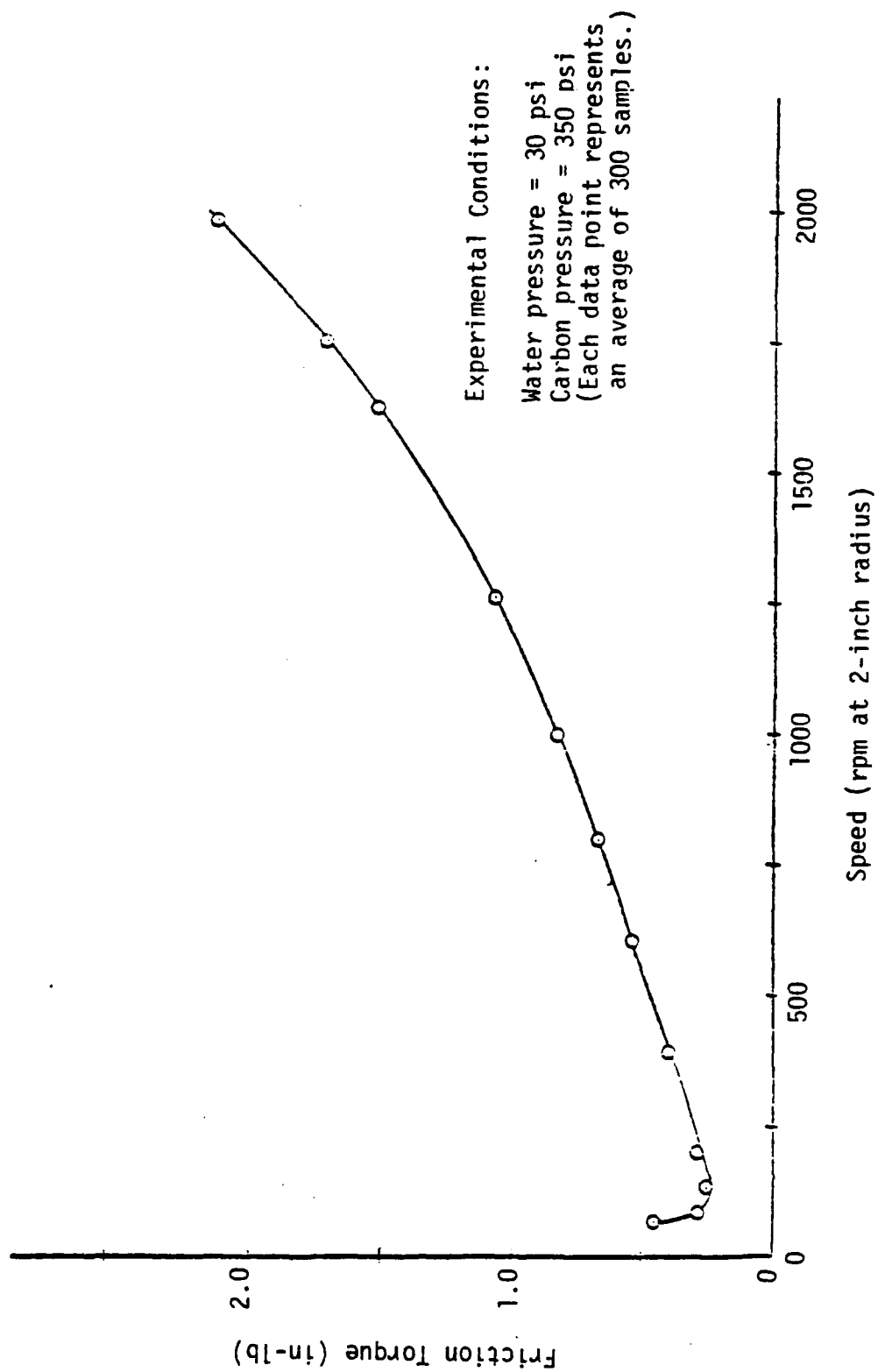


Figure 6-19. Friction Torque vs. Speed (early tests).

was very large, but stable operation could not be achieved so these data were not plotted.

Using the unmodified tester, it was also concluded that ambient pressure had no measurable effect in friction. However it was clear in these early tests that there was some uncertainty in the tests. Thus, it was at that point decided to modify the test apparatus and incorporate the hydrostatic bearing to eliminate the uncertainty caused by O-ring friction effects on the torque measurement.

After the modifications to the test apparatus, Figure 6-20 is representative of a typical test. Each point plotted represents the average of many samples. Typically 450 samples are taken during the two minute period of one test. The tests are not exactly repeatable but the trends established are.

Figure 6-21 shows friction as a function of ambient pressure at 100 and 200 rpm. First, it is to be noted that the friction is very low to start with. Second, the downward trend with increasing ambient pressure is opposite of that hypothesized. The character of the curve shown cannot be explained in terms of known theory.

It was then decided that at the higher speed the hydrodynamic effects (if they do exist) might be so strong that the modest levels of ambient pressure applied may not cause a significant decrease in cavitation. Thus tests at lower speeds were conducted by modifying the test apparatus. Figure 6-22 shows tests at very low speed. Again based on these results the hypothesis must be rejected. If anything the results show decreasing friction with increasing ambient pressure. Figure 6-23 shows the speed effect at these low speeds. It is apparent

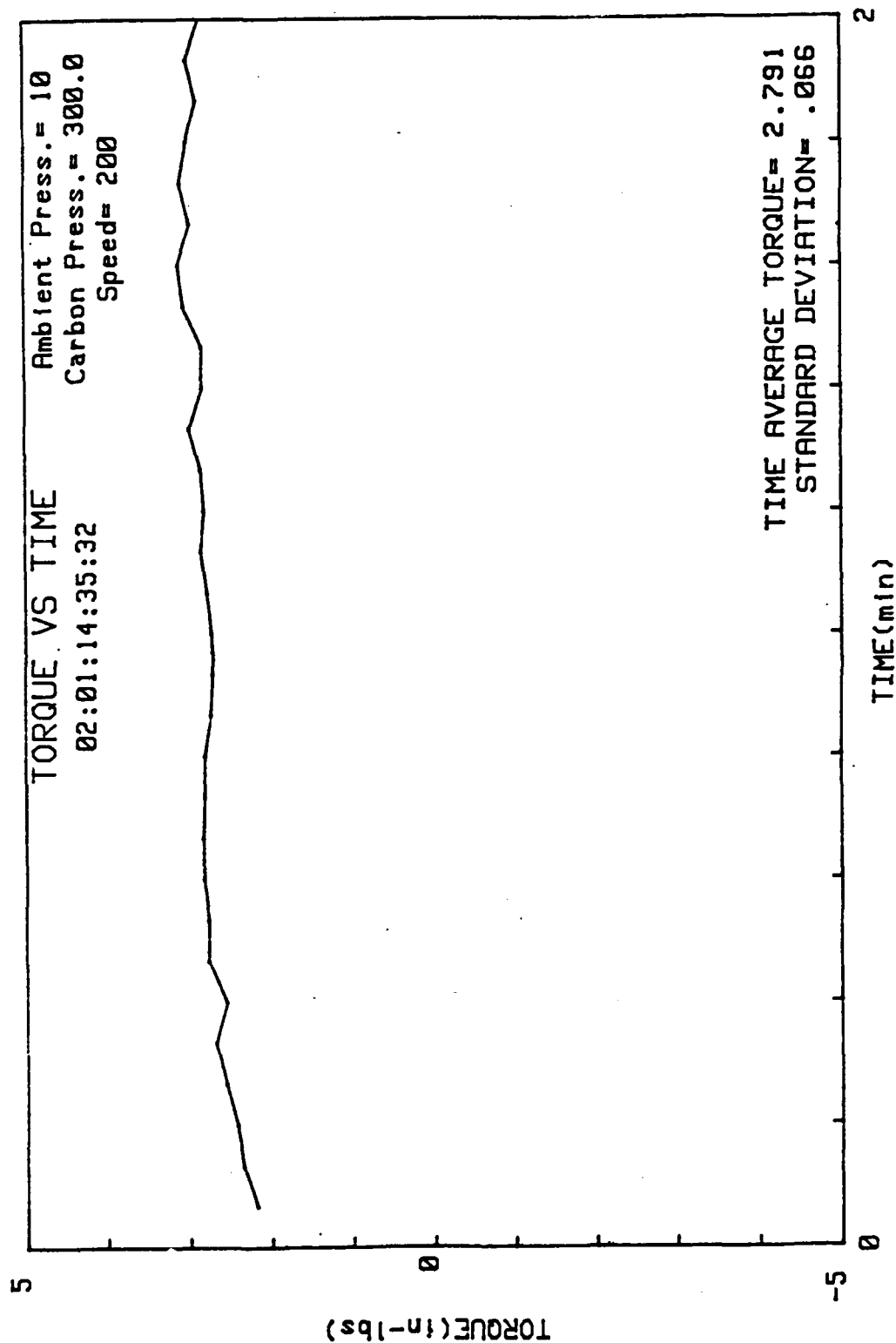


Figure 6-20. Friction Torque vs Time.

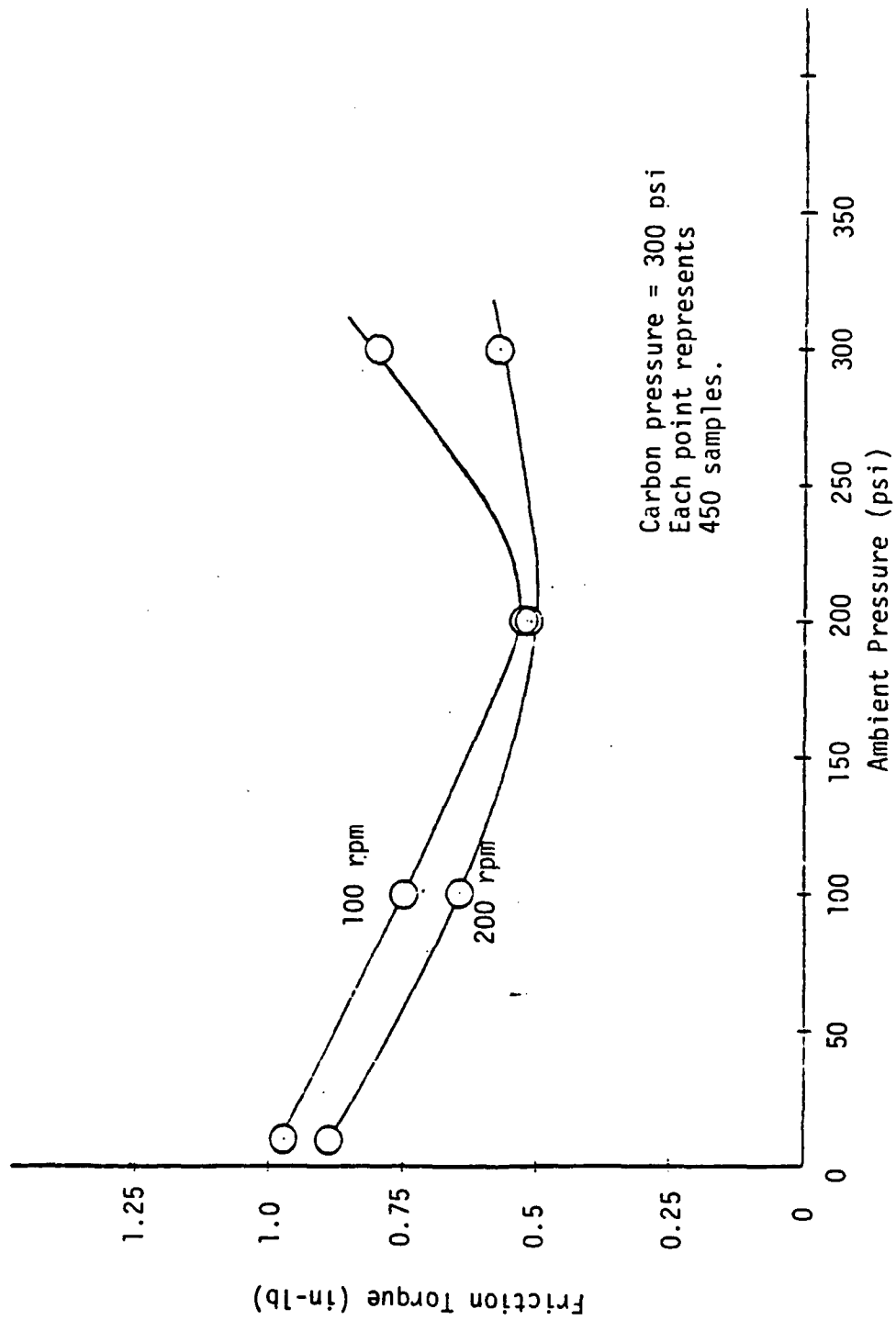


Figure 6-21. Friction Torque vs Ambient Pressure (high speed)

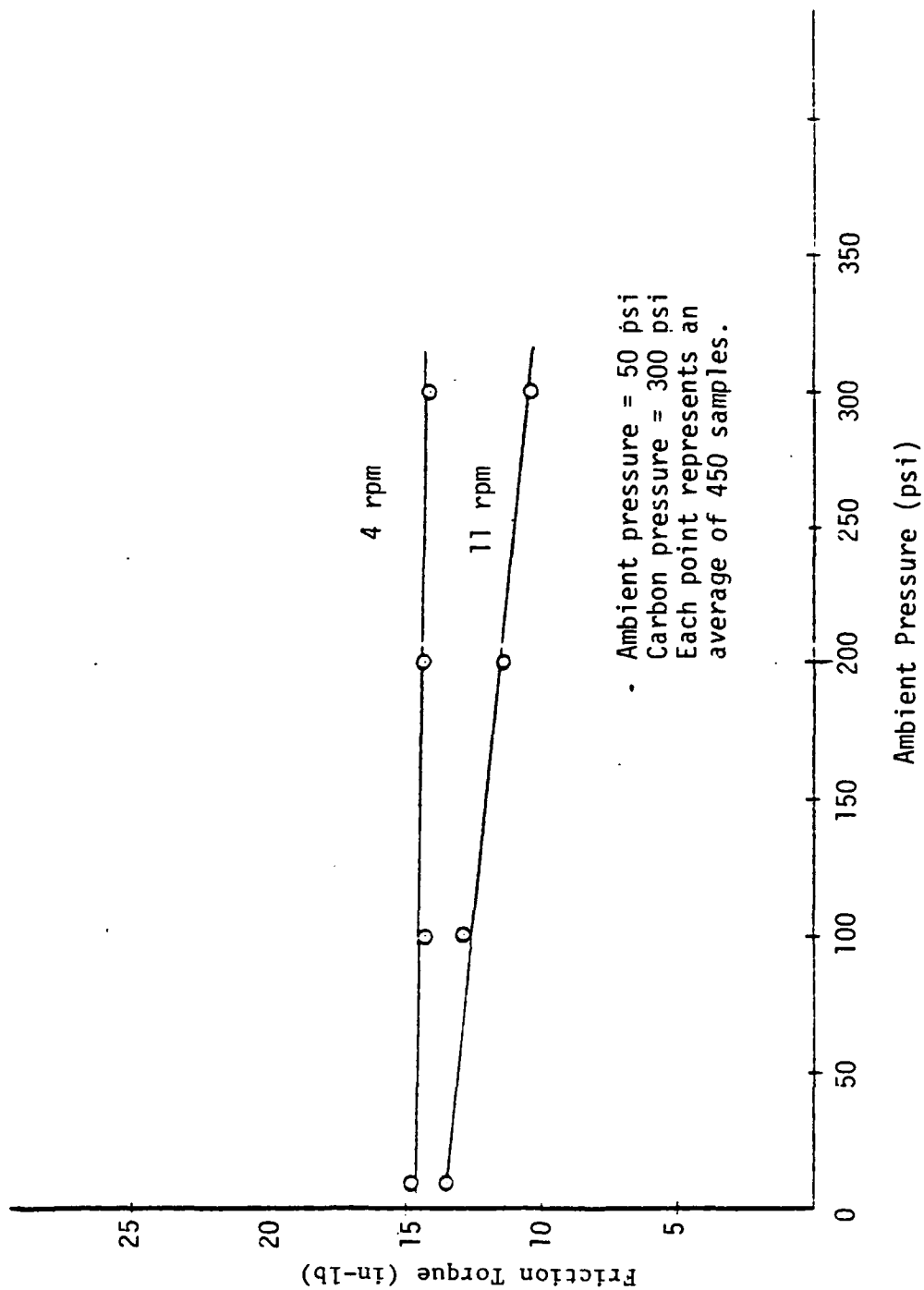


Figure 6-22. Friction Torque vs Ambient Pressure (low pressure).

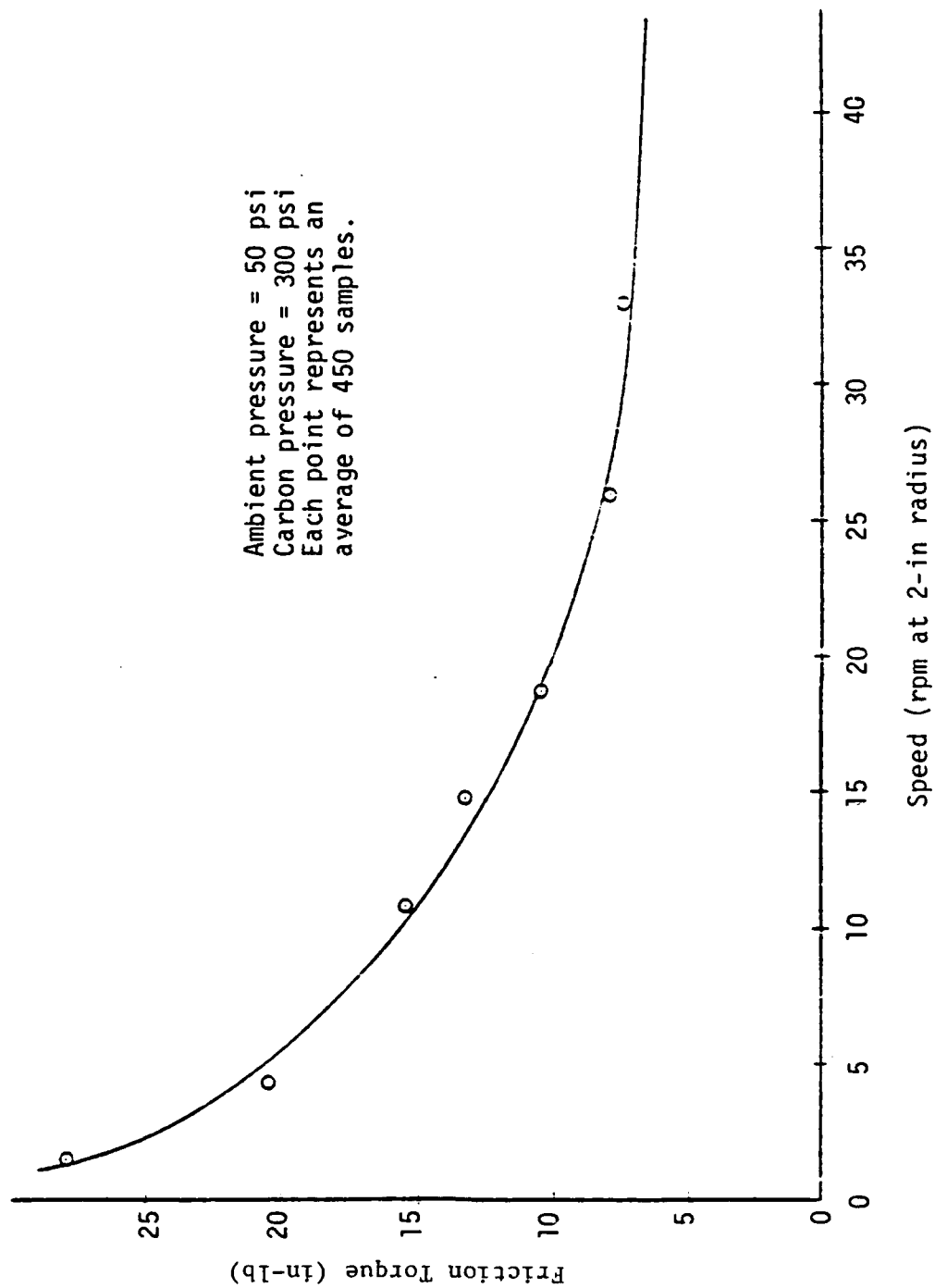


Figure 6-23. Friction Torque vs Speed (low speed).

from this curve that the speed effect was active at the speeds of the two ambient pressure tests of Figure 6-22.

The results contradicting the hypothesis were surprising. Apparently cavitation around asperities is playing no significant role in providing hydrodynamic load support in this carbon-hardface-water sliding system. Further conclusions and comments are made in Chapter 9.

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CHAPTER 7

SEAL ANALYSIS

Throughout the seals research and development effort there has been a continual need to perform numerical analysis of seals. Many computer programs have already been described in the past [1-6] which are used to analyze wavy performance and for seal design. Such programs are very specific to these unique applications and hard to generalize for others to use. There are other programs however that have possible wider application and are easier to set up for general use. These programs are described herein.

Role of Automated Analysis of Seals

Numerical tools are now available which can perform a very precise analysis of mechanical parts under load. While considerable satisfaction is gained from performing a finite element analysis of a seal ring which has already been designed. Such techniques are hard to employ during design because as soon as the smallest change is made, the mesh geometry must be entirely redefined and this is often very time consuming even if a CAD system is used. Thus to be most useful, analysis tools need to be automated as much as possible so that the designer's problem may be redefined in the simplest terms (moving a line on a screen) and the entire analysis is then made automatically.

Such a level of automation is difficult to achieve for general problems even using large software codes. However, for more

specialized problems like seals, automated analysis becomes much easier, and several such automated analysis packages have been developed herein.

Geometrical Considerations

To make it possible to do completely automated analysis for seals, advantage has been taken of certain common geometrical characteristics.

Thus for the programs that follow the following restrictions apply:

- 1) The seals are made of axisymmetric rings.
- 2) The ring cross section shape must be describable by lines parallel and perpendicular to the axis.

The second condition is the most restrictive, but most seal rings observed are either like this or can be approximated as such.

The geometry definition needed for a ring is shown in Figure 7-1. Only the corner point coordinates in clockwise order need be provided to the programs. For the example shown in the figure, 14 pairs of coordinates are needed. The starting point is arbitrary.

Once these points are input, the analysis becomes automatic. Thus, it can be seen that it is a very simple matter to change geometry and make another analysis.

Method

All of the analysis programs described require that a mesh be imposed on the cross sections for the purpose of making finite difference and finite element calculations. The essentials of this

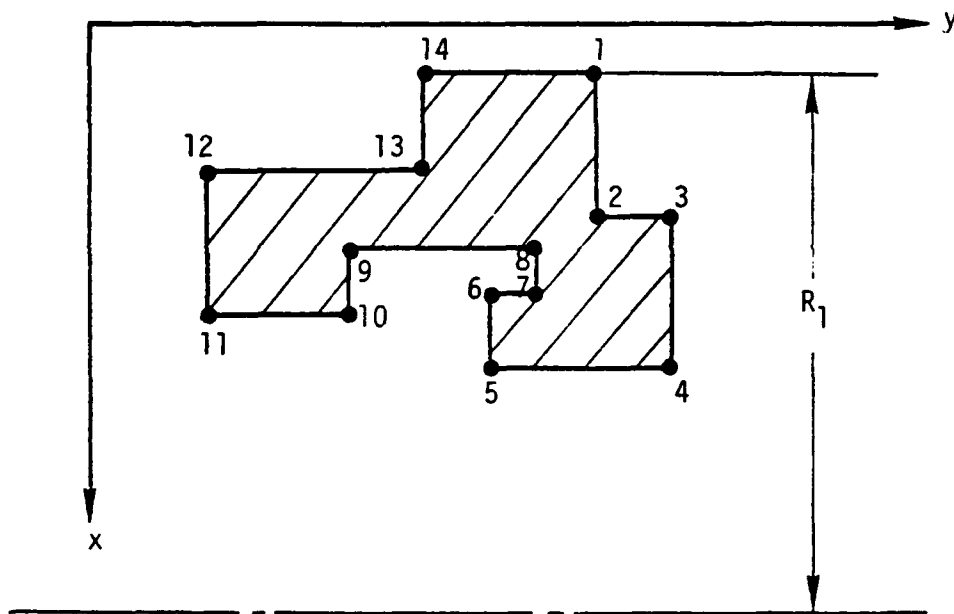


Figure 7-1. Seal Ring Cross Section Definition.

process will be described here so that the user can troubleshoot the program.

First a coarse mesh is generated based upon the sides of the section given. The coarse mesh will appear as in 7-2. The mesh will be of variable size but its boundaries will exactly coincide with the original boundaries of the part. While redundancy is eliminated by the program, having mesh points too close together is not, and the numerical consequences of such an occurrence have not been considered as yet.

After defining the coarse mesh, the program goes on to define the material contained within each rectangle of the mesh as to whether it is solid or a void. The solid-void definition is printed out as shown in the figures and will illustrate the basic shape of the object.

Once this step is completed, some computation can be performed using the coarse mesh. However, finite element, heat transfer, and torsional stiffness calculations require a refined mesh. Thus, the next step is to introduce additional mesh lines between the ones shown according to a user specified maximum D_x and D_y . Once these lines are put in, then each of the refined rectangles is defined as to solid or void and the result looks the same as Figure 7-2 but the mesh will be more uniform and finer.

At that point computations can be made. Finite difference computations are made using proper derivative approximations for the variable mesh. The finite element mesh setup simply creates rectangular elements of the size discussed. The three programs available herein are now discussed.

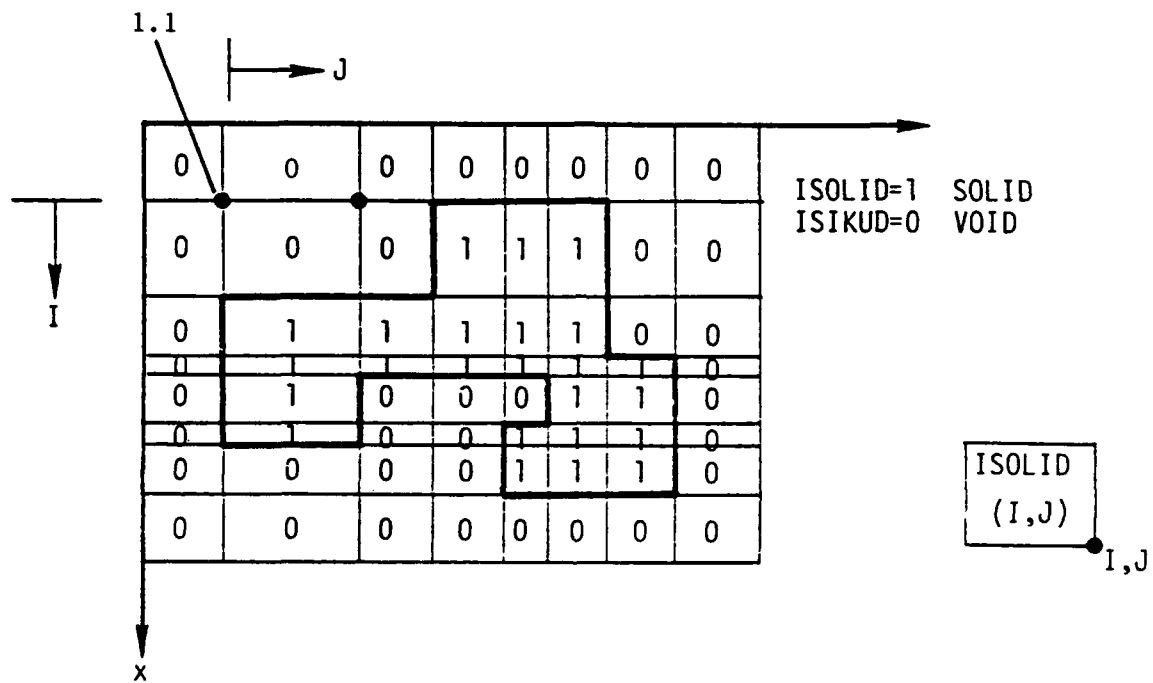


Figure 7-2. Coarse Mesh Generation

Section Properties Program

Appendix D contains the section properties program. It creates a refined mesh as described and calculates section properties.

First the centroid location is computed. Then straight beam theory moments of inertia are calculated first using an exact relationship to give

$$a = \int_A dA ,$$

$$I_x = \int_A y^2 dA ,$$

$$I_y = \int_A x^2 dA ,$$

$$I_{xy}^2 = \int_A xy dA ,$$

where x and y are in the directions as defined with the origin at the centroid. Next moments of inertia for curved beams are computed by the exact solution of the following integrals:

$$J_x = \int_A \frac{y^2}{1 - \frac{x}{R}} dA ,$$

$$J_y = \int_A \frac{x^2}{1 - \frac{x}{R}} dA ,$$

$$J_{xy} = \int_A \frac{xy}{1 - \frac{x}{R}} dA .$$

While many times the straight beam approximations are sufficiently close in value to the curved beam properties to be used, it is useful to be able to calculate both easily and compare the two. It is also useful to have J_{xy} available because J_{xy} couples in-plane and out-of-plane deflection and loads as discussed in Chapter 5.

The torsional constant is often very difficult to compute for typical seal cross sections using handbook methods. For torsion of a noncircular section, it can be shown that a Poisson's type equation applies [32]

$$\frac{\partial^2 \phi}{\partial x^2} + \frac{\partial^2 \phi}{\partial y^2} = -2 ,$$

where ϕ is a stress function. This equation was solved using a variable spacing finite difference method. Then

$$J_{\theta} = 2 \int \phi \, dA .$$

Using this program it is readily shown that the approximate formulas often used for cross sections like these are quite inaccurate.

Mesh Generation

Appendix E contains a mesh generation program. The program takes the data generated by the section program and automatically generates finite element coordinates, node numbers, element numbers, and connectivity data for rectangular axisymmetric elements to feed into a finite element program. It then goes on to ask for constraint and load input

which is properly formatted along with the above to create a data set for reading by SAPIV. This program or its equivalent is very useful to have if FEM are needed as a part of the design process.

Heat Transfer

Appendix F contains a heat transfer and thermal rotation analysis program for seal type geometries. The program permits the use of two materials with heat generation at a sliding interface between them. It also permits arbitrary convection boundaries. It solves for the temperature distributions in the two seal rings and the thermal rotation of each ring based on this temperature distribution. It uses the refined mesh generation program described in Appendix D to set up the basic mesh. Several conditions and assumptions were made when developing the program:

- 1) The program calculates the mechanical pressure at the interface assuming a linear fluid pressure drop. It then assumes that the mechanical load is uniform, that friction created by the mechanical load (per the friction coefficient) is all turned into heat, and that this results in a uniform heat generation at the seal interface.
- 2) It is assumed that both materials have the same temperature at the interface, i.e., one node is common to both materials. This is reasonable for a first approximation, but it must be recognized that in high heat flux situations contact resistance can cause the two materials to have different temperatures at their faces.

3) Thermal rotation is computed using circular ring theory [9].

It is known that this is an approximation. For more accurate results the temperature computed herein may be passed to a FEM program.

The power of this program is that it solves the arbitrary heat transfer region program by setting up finite difference equations which accommodate the variable mesh size. Boundary conditions are very easy to specify and the program sets up the appropriate heat balance equation on these boundaries as needed. Final solution is obtained very quickly by relaxation methods.

Again while this problem can be elegantly solved using FEM, the program described is completely automatic in providing a heat transfer solution, and the program is relatively small and easy to implement.

Other Programs

A program which calculates moments due to pressure forces would also be useful. The same basic program as above can be used. Pressure boundary conditions can be input similar to convection boundaries above. The program remains to be written.

All of the programs above would be easier to use if data input was done by graphical display. Thus the logical next step would be to write these programs for graphical interactive computers.

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CHAPTER 8

SQUEEZE SEALS AND BEARINGS

Introduction

Early in this research program there was motivation to find a simple means of applying waviness to seals to more easily obtain the advantages of waviness. As discussed earlier, while the advantages of using waviness have been clear, a very simple means of applying the wave is essential to make the idea practical.

The squeeze seal concept was born of this motivation. Figure 8-1 shows this concept. Here a wave moves along a stationary (nonsliding) surface on one part at velocity U_2 . The second part slides along in a conventional manner (velocity U_1). Of interest is the case where $U_1 = U_2$ because it is easy to conceive how this might be simply implemented.

The next section shows how this problem is solved and how in fact the squeeze seal/bearing is identical to the standard wavy seal/bearing (Figure 8-2). Thus, while the squeeze bearing may provide an advantage in implementation, it does not provide an advantage in performance. The squeeze seal concept then led to the third wavy seal concept described in Chapter 4 where the wave moves with the rotating part of the seal.

While it happens that the above three concepts are identical in performance, there is yet another way of creating a wavy squeeze seal of some interest. The concept is shown in Figure 8-3.

Surface A is a conventional wavy surface if it is used as a seal and may be a multilobed journal if used as a bearing. Surface B is a

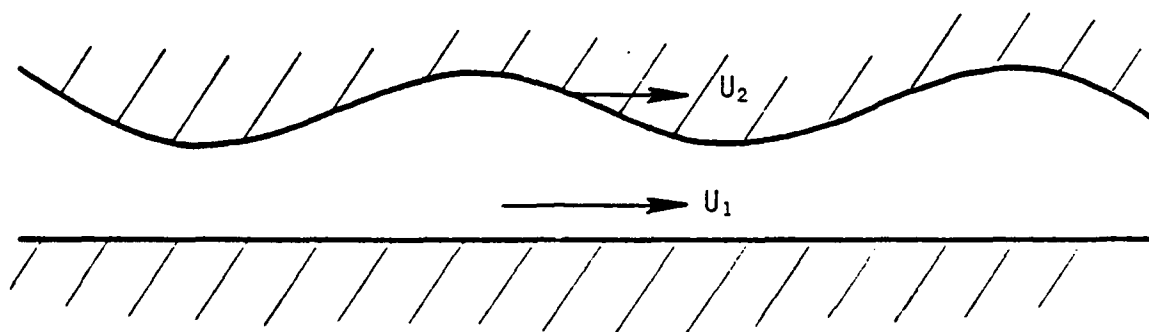


Figure 8-1. Squeeze Seal/Bearing

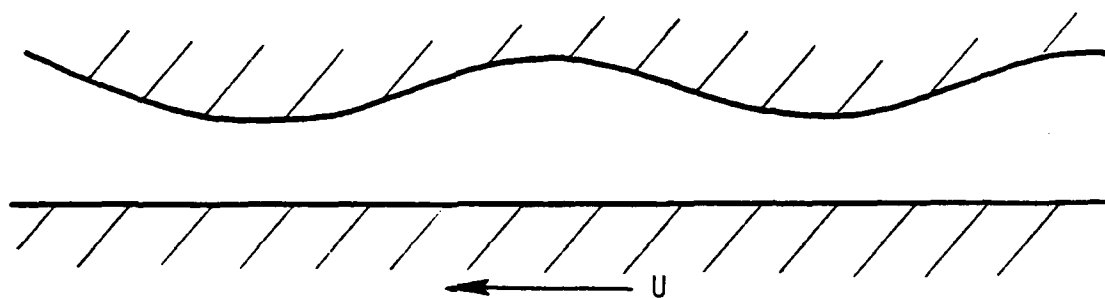


Figure 8-2. Conventional Wavy Seal/Bearing.

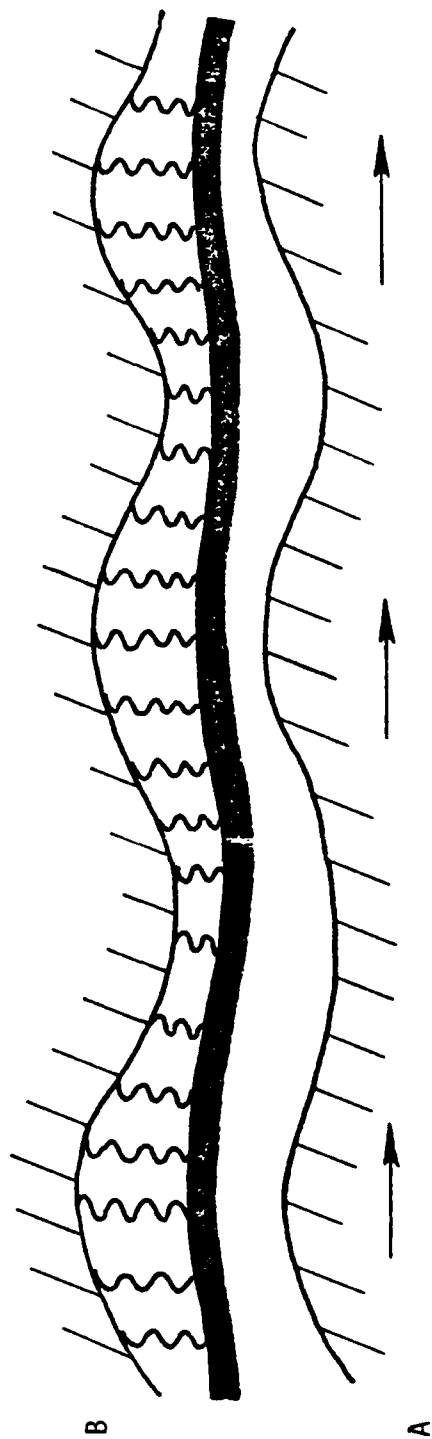


Figure 8-3. Elastic Foundation Squeeze Seal/Bearing.

beam or plate on an elastic foundation which has a variable stiffness support along its length. Thus, as surface A slides along, surface B will react differently at different locations because of the variable stiffness. The analysis and performance of this concept is discussed in a later section.

Rigid Body Squeeze Bearings/Seals

In this section, the governing finite difference equations for both conventional and squeeze bearings/seals are first derived by using Elrod's cavitation algorithm [53]. The techniques for obtaining values of pressure from the governing equation accompanied by the given boundary conditions are presented and discussed. Although the governing equations for both conventional and squeeze bearings are different, it can be proven through analytical methods that performance for both bearings/seals under complete film condition is exactly the same. For conditions in which cavitation occurs, it can also be proven, by using numerical methods, that performance for both bearings/seals is identical.

Governing Equations

Assumptions

The assumptions upon which the present solutions are based are as follows:

- 1) Lubricant is Newtonian
- 2) Flow is laminar

- 3) Fluid film is so thin that the pressure remains constant across its depth.
- 4) No slip occurs between fluid and bearing/seal surfaces.
- 5) The viscosity is constant throughout the film.
- 6) The effects of thermal and elastic distortion are neglected.
- 7) Inertia and body forces are negligible.
- 8) The curvature of surfaces is large compared with film thickness.

The justifications for these assumptions have been presented previously and will not be repeated here.

According to Cameron [54], with these assumptions the flow rate in the x direction per unit y width is

$$q_x = - \frac{h^3}{12\mu} \frac{\partial p}{\partial x} + \frac{h}{2} (U_a + U_b) . \quad (8-1)$$

Similarly, in the y direction

$$q_y = - \frac{h^3}{12\mu} \frac{\partial p}{\partial y} + \frac{h}{2} (V_a + V_b) , \quad (8-2)$$

where U_a , U_b are the surface velocities in x direction, V_a , V_b are the surface velocities in the y direction. The continuity equation can be described as

$$\frac{\partial q_x}{\partial x} + \frac{\partial q_y}{\partial y} = \frac{\partial h}{\partial t} . \quad (8-3)$$

Equations (8.1), (8.2), and (8.3) can only be used in a complete film zone. The problem of cavitation must be dealt with. The cavitation phenomenon is caused by negative pressure occurring in a fluid

film lubrication region. It was found by Jacobson et al. [55] that in a cavitation zone, only a certain fraction, θ , of the film gap is occupied by the fluid and the remaining space, $1 - \theta$, is occupied by the intervening gas. The pressure throughout the cavitation zone is constant.

By using Jacobson's theory, Elrod [53] modified the Reynolds equation and proposed a computational method which will be reviewed here. Elrod defines the fractional film content, θ , as

$$\theta = \rho / \rho_c , \quad (8-4)$$

where ρ_c is the liquid density within the cavitated zone and ρ is the average density. The film pressure is

$$\begin{aligned} p &= p_c + \beta(\theta - 1) , & \theta > 1 , \\ p &= p_c , & \theta \leq 1 , \end{aligned} \quad (8-5)$$

where p_c is the cavity pressure and β is the compression factor (for an essentially incompressible problem, a convenient arbitrary value can be used).

A cavitation index g is also defined as

$$\begin{aligned} g &= 0 \quad \text{for } \theta < 1 , \\ g &= 1 \quad \text{for } \theta \geq 1 . \end{aligned} \quad (8-6)$$

By using the defined variables and finite difference method (nomenclature shown in Figure 8-4), Equation (8-1) was modified as [53]

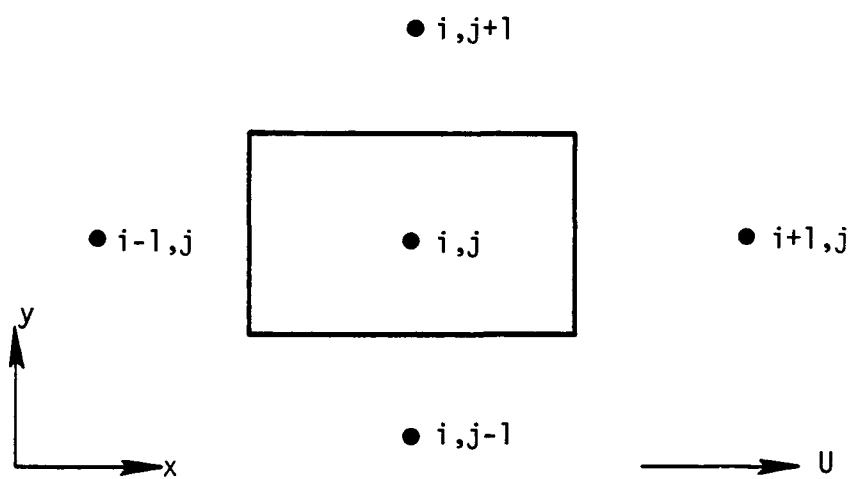


Figure 8-4. Finite Difference Symbol Definition.

$$\begin{aligned}
q_x = & \left(\frac{h^3}{12\mu} \right)_{av} \rho_c \left\{ \frac{g_{i-1,j}(\theta_{i-1,j} - 1) - g_{i,j}(\theta_{i,j} - 1)}{\Delta x} \right\} \\
& + \frac{\rho_c U}{2} \left\{ \theta_{i-1,j} h_{i-1,j} (1 - g_{i-1,j}) + g_{i-1,j} h_{i-1,j} \right. \\
& \left. + \frac{g_{i,j} g_{i-1,j} (h_{i,j} - h_{i-1,j})}{2} \right\} \quad (8-7)
\end{aligned}$$

where U is the surface velocity in the x direction. Equation (8-2) was modified in a similar way.

Based on the cavitation concept, Equation (8-3) is also modified to

$$\frac{\Delta q_x}{\Delta x} + \frac{\Delta q_y}{\Delta y} = \frac{\partial [h(\theta(1 - g) + g)]}{\partial t} \quad (8-8)$$

Equations for Conventional and Squeeze Wavy Seals With No Cavitation

For the complete film condition, Equation (8-3) can be used. The derivation of the governing equations is as follows (for the present problems, only one of the surfaces of bearing/seal has a constant moving velocity:

$$\frac{\partial q_x}{\partial x} = \frac{\partial}{\partial x} \left(-\frac{h^3}{12\mu} \frac{\partial p}{\partial x} + \frac{h}{2} U \right) = -\frac{1}{12\mu} \frac{\partial}{\partial x} \left(h^3 \frac{\partial p}{\partial x} \right) + \frac{U}{2} \frac{2h}{\partial x}, \quad (8-9)$$

$$\frac{\partial q_y}{\partial y} = \frac{\partial}{\partial y} \left(-\frac{h^3}{12\mu} \frac{\partial p}{\partial y} \right) = -\frac{1}{12\mu} \frac{\partial}{\partial y} \left(h^3 \frac{\partial p}{\partial y} \right). \quad (8-10)$$

Substitute Equations (8-9) and (8-10) into Equation (8-3) and rearrange its terms, Equation (8-3) becomes (for conventional wavy seal)

$$\frac{\partial}{\partial x} \left(h^3 \frac{\partial p}{\partial x} \right) + \frac{\partial}{\partial y} \left(h^3 \frac{\partial p}{\partial y} \right) = 6\mu U \frac{\partial h}{\partial x} \quad (8-11)$$

For squeeze wavy seal:

$$\frac{\partial}{\partial x} \left(h^3 \frac{\partial p}{\partial x} \right) + \frac{\partial}{\partial y} \left(h^3 \frac{\partial p}{\partial y} \right) = 6\mu \left(U \frac{\partial h}{\partial x} - 2 \frac{\partial h}{\partial t} \right) \quad (8-12)$$

Equations for Conventional and Squeeze Wavy Seals with Cavitation

For conditions in which cavitation occurs, Equation (8-3) alone is not enough to fully represent the fluid film performance and Equation (8-8) is used. By using the Elrod's cavitation algorithm, the governing equations for a rigid squeeze wavy seal with cavitation are obtained:

$$\begin{aligned} \frac{\Delta q_x}{\Delta x} &= \frac{1}{\Delta x} \left[q_{i-1/2 \cdot j \cdot t} - q_{i+1/2 \cdot j \cdot t} \right] \\ &= \frac{\beta p_c}{12\mu} \left[\left(\frac{h_{i-1 \cdot j \cdot t} + h_{i \cdot j \cdot t}}{2} \right)^3 \left(\frac{g_{i-1 \cdot t}(\theta_{i-1 \cdot j \cdot t} - 1) - g_{i \cdot j \cdot t}(\theta_{i \cdot j \cdot t} - 1)}{(\Delta x)^2} \right) \right. \\ &\quad \left. - \left(\frac{h_{i \cdot j \cdot t} + h_{i+1 \cdot j \cdot t}}{2} \right)^3 \left(\frac{g_{i \cdot j \cdot t}(\theta_{i \cdot j \cdot t} - 1) - g_{i+1 \cdot j \cdot t}(\theta_{i+1 \cdot j \cdot t} - 1)}{(\Delta x)^2} \right) \right] \\ &\quad + \frac{U \rho_c}{2\Delta x} \left[\theta_{i-1 \cdot j \cdot t} h_{i-1 \cdot j \cdot t} (1 - g_{i-1 \cdot j \cdot t}) + g_{i-1 \cdot j \cdot t} h_{i-1 \cdot j \cdot t} \right. \\ &\quad \left. + \frac{1}{2} q_{i \cdot j \cdot t} q_{i-1 \cdot j \cdot t} (h_{i \cdot j \cdot t} - h_{i-1 \cdot j \cdot t}) \right] \end{aligned}$$

$$\begin{aligned}
& - \frac{U\rho_c}{2\Delta x} \left[\theta_{i,j,t} h_{i,j,t} (1 - g_{i,j,t}) + g_{i,j,t} h_{i,j,t} \right. \\
& \quad \left. + \frac{1}{2} g_{i+1,j,t} g_{i,j,t} (h_{i+1,j,t} - h_{i,j,t}) \right] .
\end{aligned} \tag{8-13}$$

$$\begin{aligned}
\frac{\Delta q_y}{\Delta y} &= \frac{1}{\Delta y} [q_{i,j-1/2,t} - q_{i,j+1/2,t}] \\
&= \frac{\beta\rho_c}{12\mu} \left[\left(\frac{h_{i,j-1,t} + h_{i,j,t}}{2} \right)^3 \left(\frac{g_{i,j-1,t}(\theta_{i,j-1,t} - 1) - g_{i,j,t}(\theta_{i,j,t} - 1)}{(\Delta y)^2} \right) \right. \\
&\quad \left. - \left(\frac{h_{i,j,t} + h_{i,j+1,t}}{2} \right)^3 \left(\frac{g_{i,j,t}(\theta_{i,j,t} - 1) - g_{i,j+1,t}(\theta_{i,j+1,t} - 1)}{(\Delta y)^2} \right) \right]
\end{aligned} \tag{8-14}$$

$$\begin{aligned}
& \frac{\partial(h(\theta(1 - g) + g))}{\partial t} \\
&= \frac{h_{i,j,t}(\theta_{i,j,t}(1 - g_{i,j,t}) + g_{i,j,t})}{\Delta t} \\
&\quad - \frac{h_{i,j,t-\Delta t}(\theta_{i,j,t-\Delta t}(1 - g_{i,j,t-\Delta t}) + g_{i,j,t-\Delta t})}{\Delta t}
\end{aligned} \tag{8-15}$$

By substituting Equation (8-13), (8-14), and (8-15) into Equation (8-8), the governing equation for a two-dimensional squeeze wavy seal can be obtained.

The governing equation for a one-dimensional squeeze wavy seal is:

$$\theta_{i+1,t} \left[- \frac{\beta\rho_c}{12\mu\Delta x} \left(\frac{h_{i,t} + h_{i+1,t}}{2} \right)^3 g_{i+1,t} \right]$$

$$\begin{aligned}
& + \theta_{i-1,t} \left[- \frac{8\rho_c}{12\mu\Delta x} \left(\frac{h_{i,t} + h_{i-1,t}}{2} \right)^3 g_{i-1,t} - \rho_c \frac{U}{2} h_{i-1,t} (1 - g_{i-1,t}) \right] \\
& + \theta_{i,t} \left[\frac{8\rho_c}{12\mu\Delta x} \left[\left(\frac{h_{i,t} + h_{i+1,t}}{2} \right)^3 g_{i,t} + \left(\frac{h_{i,t} + h_{i-1,t}}{2} \right)^3 g_{i-1,t} \right] \right. \\
& \quad \left. + \rho_c \frac{U}{2} h_{i,t} (1 - g_{i,t}) + \rho_c \frac{\Delta x}{\Delta t} h_{i,t} (1 - g_{i,t}) \right] \\
& = \frac{8\rho_c}{12\mu\Delta x} \left[\left(\frac{h_{i,t} + h_{i-1,t}}{2} \right)^3 (g_{i,t} - g_{i-1,t}) \right. \\
& \quad \left. + \left(\frac{h_{i,t} + h_{i+1,t}}{2} \right)^3 (g_{i,t} - g_{i+1,t}) \right] \\
& + \rho_c \frac{U}{2} [g_{i-1,t} h_{i-1,t} - g_{i,t} h_{i,t} \\
& \quad + 0.5 g_{i,t} g_{i-1,t} (h_{i,t} - h_{i-1,t}) \\
& \quad - 0.5 g_{i+1,t} g_{i,t} (h_{i+1,t} - h_{i,t})] \\
& + \rho_c \frac{\Delta x}{\Delta t} [h_{i,t-1} g_{i,t-1} - h_{i,t} g_{i,t}] \\
& + \theta_{i,t-1} \left[\rho_c \frac{\Delta x}{\Delta t} h_{i,t-1} (1 - g_{i,t-1}) \right]. \tag{8-16}
\end{aligned}$$

For a conventional wavy seal,

$$\frac{\partial(h(\theta(1-g) + g))}{2t} \tag{8-15}$$

is equal to zero (due to steady state condition). The governing equation can be obtained by combining Equation (8-13) and (8-14).

Comparison Between Conventional and Rigid Squeeze Wavy Bearings/Seals

(a) Complete film condition occurs throughout the bearing.

Under complete film conditions the governing equation for a squeeze wavy bearing/seal is

$$\frac{\partial}{\partial x} \left(h^3 \frac{\partial p}{\partial x} \right) + \frac{\partial}{\partial y} \left(h^3 \frac{\partial p}{\partial y} \right) = 6\mu U \frac{\partial h}{\partial x} - 12\mu \frac{\partial h}{\partial t} . \quad (8-12)$$

Due to the fact from Figure 8-1 with

$$U_1 = U_2 = U ,$$

$$\frac{\partial h}{\partial t} = -U \frac{\partial h}{\partial x} ,$$

the right-hand side terms of Equation (8-12) can be expressed as

$$6\mu U \frac{\partial h}{\partial x} - 12\mu \frac{\partial h}{\partial t} = -6\mu U \frac{\partial h}{\partial x}$$

and Equation (8-12) becomes

$$\frac{\partial}{\partial x} \left(h^3 \frac{\partial p}{\partial x} \right) + \frac{\partial}{\partial y} \left(h^3 \frac{\partial p}{\partial y} \right) = 6\mu(-U) \frac{\partial h}{\partial x} \quad (8-17)$$

Since the governing equation for a conventional wavy seal is

$$\frac{\partial}{\partial x} \left(h^3 \frac{\partial p}{\partial x} \right) + \frac{\partial}{\partial y} \left(h^3 \frac{\partial p}{\partial y} \right) = 6\mu U \frac{\partial h}{\partial x} . \quad (8-10)$$

It is obvious that a squeeze wavy seal can be treated as a conventional wavy seal except that now the moving surface is moved in an opposite direction. Actually, this does not effect the performance. Numerical results for Equations (8-10) and (8-17) for a two-dimensional region

with the film thickness shown by Table 8-1 are shown in Tables 8-2 and 8-3. They are identical except for the moving direction.

(b) Cavitation occurs somewhere in the fluid film

If cavitation occurs, the governing equations for both conventional and squeeze wavy seals are tedious. The only way to solve these equations is through iterative computational methods.

The following are two examples; one deals with the one-dimensional case and the other two-dimensional case. The comparison between conventional and squeeze wavy seals/bearings is considered:

(i) One-dimensional Case

In this one-dimensional problem, the governing equation for a squeeze wavy seal is given as Equation (8-16) and the film thickness is

$$h = h_0 + h_1 \cos(n(x - Ut)) , \quad (8-18)$$

where

h_0 = nominal film thickness

h_1 = amplitude of wave

n = number of waves around seal/bearing face

t = time

The boundary conditions for use here are described as follows:

- 1) At the beginning of each wave (each wave begins with point where film thickness is a maximum), the pressure is assumed to be zero. This means these points are moved with the wave.

TABLE 8-1

Distribution of Film Thickness over One Wave of a Seal Face

		r							
	θ	1	2	3	4	5	6	7	8
	1	1.23	1.74	2.35	2.97	3.58	4.20	4.82	5.43
	2	1.23	1.73	2.33	2.94	3.54	4.15	4.75	5.36
	3	1.24	1.70	2.28	2.85	3.43	4.00	4.58	5.15
	4	1.25	1.66	2.19	2.71	3.24	3.76	4.29	4.82
	5	1.26	1.61	2.07	2.53	3.00	3.46	3.92	4.38
	6	1.27	1.55	1.94	2.32	2.71	3.10	3.49	3.87
	7	1.29	1.48	1.79	2.10	2.41	2.71	3.02	3.33
	8	1.30	1.42	1.65	1.87	2.10	2.33	2.56	2.79
	9	1.31	1.36	1.51	1.66	1.82	1.97	2.13	2.28
	10	1.32	1.30	1.39	1.48	1.57	1.66	1.76	1.85
	11	1.33	1.26	1.31	1.35	1.39	1.43	1.47	1.51
	12	1.34	1.24	1.25	1.26	1.27	1.28	1.29	1.30
	13	1.34	1.23	1.23	1.23	1.23	1.23	1.23	1.23
	14	1.34	1.24	1.25	1.26	1.27	1.28	1.29	1.30
	15	1.33	1.26	1.31	1.35	1.39	1.43	1.47	1.51
	16	1.32	1.30	1.39	1.48	1.57	1.66	1.76	1.85
	17	1.31	1.36	1.51	1.66	1.82	1.97	2.13	2.28
	18	1.30	1.42	1.65	1.87	2.10	2.33	2.56	2.79
	19	1.29	1.48	1.79	2.10	2.41	2.71	3.02	3.33
	20	1.27	1.55	1.94	2.32	2.71	3.10	3.49	3.87
	21	1.26	1.61	2.07	2.53	3.00	3.46	3.92	4.38
	22	1.25	1.66	2.19	2.71	3.24	3.76	4.29	4.82
	23	1.24	1.70	2.28	2.85	3.43	4.00	4.58	5.15
	24	1.23	1.73	2.33	2.94	3.54	4.15	4.75	5.36
	25	1.23	1.74	2.35	2.97	3.58	4.20	4.82	5.43

TABLE 8-2

Pressure Distribution for a Conventional Wavy Seal
under Full Film Conditions

U = 0.1885E+03 U1 = 0.1885E+03 VO = 0.7200E-05
P2 = 0.4000E+04 CO = 0.2000E-84
HO = 0.2460E-84 FO = 0.2299E-03

		r							
θ	1	0	112	155	174	184	190	194	197
	2	0	115	158	177	187	192	195	197
	3	0	116	161	180	189	194	196	197
	4	0	116	163	183	192	196	197	197
	5	0	114	164	186	196	199	199	197
	6	0	111	166	190	200	203	201	197
	7	0	108	167	195	207	209	205	197
	8	0	102	167	200	215	216	210	197
	9	0	94	163	204	223	226	216	197
	10	0	83	153	200	226	233	223	197
	11	0	68	133	182	214	228	222	197
	12	0	49	100	144	177	198	206	197
	13	0	30	62	93	121	148	173	197
	14	0	14	30	49	73	103	143	197
	15	0	6	16	31	54	88	134	197
	16	0	8	20	39	65	99	143	197
	17	0	19	38	62	89	120	156	197
	18	0	34	62	88	114	141	168	197
	19	0	50	85	112	135	157	177	197
	20	0	66	105	131	151	168	183	197
	21	0	81	121	145	163	176	187	197
	22	0	92	134	156	171	181	190	197
	23	0	101	143	164	176	185	192	197
	24	0	108	150	169	181	188	193	197
	25	0	112	155	174	184	190	194	197

The load support is 0.7355E+03.

TABLE 8-3

Pressure Distribution for a Squeeze Wavy Seal Seal
under Full Film Conditions

		r							
θ	1	0	112	155	174	184	190	194	197
	2	0	108	150	169	181	188	193	197
	3	0	101	143	164	176	185	192	197
	4	0	92	134	156	171	181	190	197
	5	0	81	121	145	163	176	187	197
	6	0	66	105	131	151	168	183	197
	7	0	50	85	112	135	157	177	197
	8	0	34	62	88	114	141	168	197
	9	0	19	38	62	89	120	156	197
	10	0	8	20	39	65	99	143	197
	11	0	6	16	31	54	88	134	197
	12	0	14	30	49	73	103	143	197
	13	0	30	62	93	121	148	173	197
	14	0	49	100	144	177	198	206	197
	15	0	68	133	182	214	228	222	197
	16	0	83	153	200	226	233	223	197
	17	0	94	163	204	223	226	216	197
	18	0	102	167	200	215	216	210	197
	19	0	108	167	195	207	209	205	197
	20	0	111	166	190	200	203	201	197
	21	0	114	164	186	196	199	199	197
	22	0	116	163	183	192	196	197	197
	23	0	116	161	180	189	194	196	197
	24	0	115	158	177	187	192	195	197
	25	0	112	155	174	184	190	194	197

The load support is 0.7355E+03.

- 2) For each wave, at a location where cavitation first occurs, the pressure and pressure gradient are equal to zero.

The reason why these two boundary conditions are chosen actually comes from the idea that a lubricant film having converging-diverging geometry (such as a journal bearing) usually increases its pressure from the beginning of its convergent section. At the location where cavitation first occurs, pressure and pressure gradient are equal to zero.

Although these two boundary conditions are actually used to obtain the solution, only the first boundary condition is included directly in the solution procedure. The reason is that the governing equation used here is derived from Elrod's cavitation algorithm and this algorithm itself has the ability to automatically handle the cavitation boundary condition.

To find the solution, the following procedures are used.

- (1) All $g_{i,t}(t = 0)$ were set equal to one at the start.
- (2) Value of $\theta_{i,t}$ at the beginning of each wave was obtained by Equation (8-5) using an assumed pressure distribution.
- (3) By using the Gauss elimination method, the distribution of $\theta_{i,t}(t = 0)$ was obtained.
- (4) By using Equation (6), new $g_{i,t}(t = 0)$ were found.
- (5) Using new $g_{i,t}(t = 0)$ and repeat procedure (2)-(4), until the solutions of $\theta_{i,t}(t = 0)$ are converged.
- (6) Assume the values of all $g_{i,t+\Delta t}$ were equal to 1.
- (7) Use Gauss elimination method to find $\theta_{i,t+\Delta t}$.
- (8) By using Equation (8-6), new $g_{i,t+\Delta t}$ were found.

- (9) The new $g_{i,t+\Delta t}$ were used and repeat procedure (7)-(9) until the solution of $\theta_{i,t+\Delta t}$ were converged.
- (10) By using Equation (5) and assuming all negative pressures are equal to zero, the pressure distribution was found.

In all the iteration processes, the criterion to decide whether the solution converges is that three successively determined load supports should not differ by more than 0.1 percent.

The method used to solve the governing equation for a conventional wavy seal/bearing is similar to that of a squeeze wavy seal/bearing except that now the film thickness is

$$h = h_0 + h_1 \cos nx . \quad (8-19)$$

The numerical data used here are

$$h_0 = 0.383 \times 10^{-3} \text{ in.}$$

$$h_1 = 0.3 \times 10^{-4} \text{ in.}$$

$$n = 3$$

$$\beta = 5000$$

$$U = 377 \text{ in./sec}$$

$$\mu = 0.99 \times 10^{-7} \text{ Reyn.}$$

The numerical results obtained for the conventional and squeeze wavy seals are not exactly the same (see Figure 8-5). However, it was found that by reducing the size of the mesh, the value of the load support for a squeeze wavy seal converges to that of a conventional wavy seal (see Figure 8-6). This proves that if the mesh size is small enough, the results for both conventional and squeeze wavy seals are identical.

One wave is divided into 84 sections.

- * Conventional Wavy Seal
- o Squeeze Wavy Seal

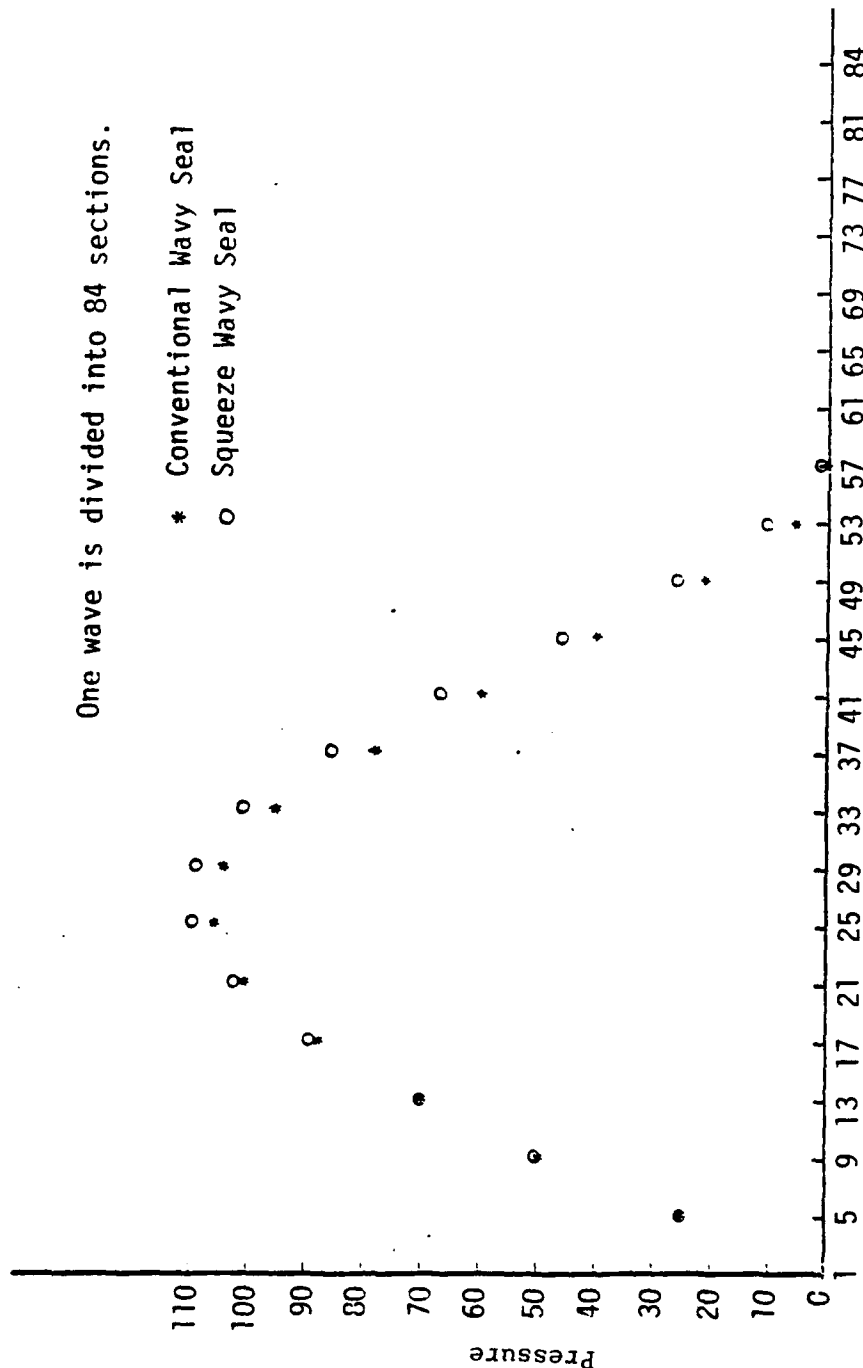


Figure 8-5. Comparison of Pressure with Cavitation.

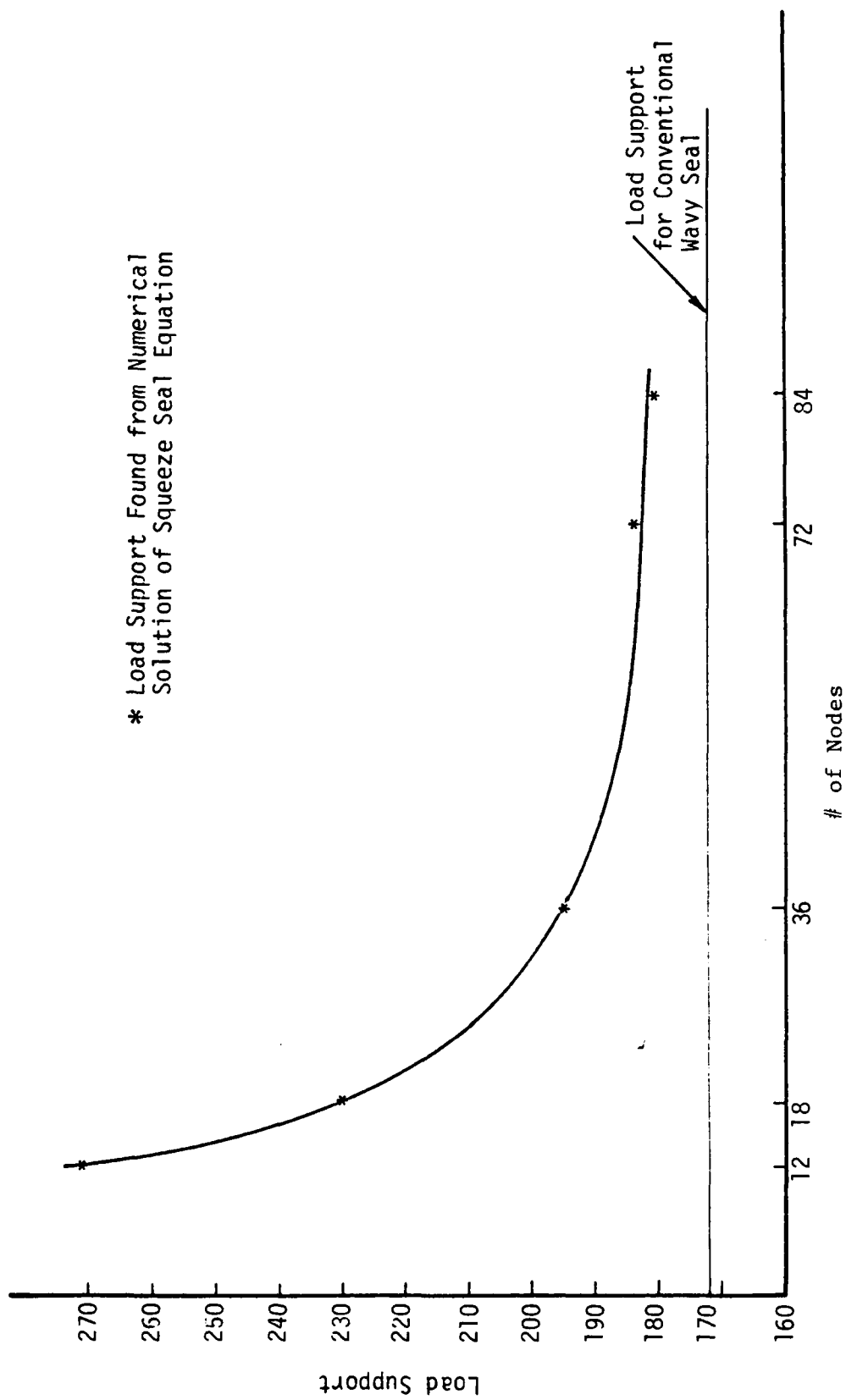


Figure 8-6. Effect of Mesh Size on Squeeze Seal Load Support.

(ii) Two-dimensional Case

Consider a two-dimensional squeeze film flow occurring between two circular rings (seal). The coordinates here are cylindrical. The nomenclature is shown in Figure 8-7.

The derivation procedure of the governing equation for the present problem is similar to that for rectangular coordinates obtained in the previous section. This governing equation is given as

$$\begin{aligned}
 & \theta_{i+1,j,t} \left[- \frac{\beta \rho_c \Delta r}{12 \mu r \Delta \alpha} \left(\frac{h_{i,j,t} + h_{i+1,j,t}}{2} \right)^3 g_{i+1,j,t} \right] \\
 & + \theta_{i-1,j,t} \left[- \frac{\beta \rho_c \Delta r}{12 \mu r \Delta \alpha} \left(\frac{h_{i-1,j,t} + h_{i,j,t}}{2} \right)^3 g_{i-1,j,t} \right. \\
 & \quad \left. - \frac{r \omega \Delta r}{2} h_{i-1,j,t} (1 - g_{i-1,j,t}) \right] \\
 & + \theta_{i,j+1,t} \left[- \frac{\beta \rho_c (r + \frac{\Delta r}{2}) \Delta \alpha}{12 \mu \Delta r} \left(\frac{h_{i,j,t} + h_{i,j+1,t}}{2} \right)^3 g_{i,j+1,t} \right] \\
 & + \theta_{i,j-1,t} \left[- \frac{\beta \rho_c (r - \frac{\Delta r}{2}) \Delta \alpha}{12 \mu \Delta r} \left(\frac{h_{i,j,t} + h_{i,j-1,t}}{2} \right)^3 g_{i,j-1,t} \right] \\
 & + \theta_{i,j,t} \left[\frac{\beta \rho_c \Delta r}{12 \mu r \Delta \alpha} \left(\frac{h_{i,j,t} + h_{i+1,j,t}}{2} \right)^3 g_{i,j,t} \right. \\
 & \quad + \frac{\beta \rho_c \Delta r}{12 \mu r \Delta \alpha} \left(\frac{h_{i-1,j,t} + h_{i,j,t}}{2} \right)^3 g_{i,j,t} \\
 & \quad \left. + \frac{\beta \rho_c (r + \frac{\Delta r}{2}) \Delta \alpha}{12 \mu \Delta r} \left(\frac{h_{i,j,t} + h_{i,j+1,t}}{2} \right)^3 g_{i,j,t} \right]
 \end{aligned}$$

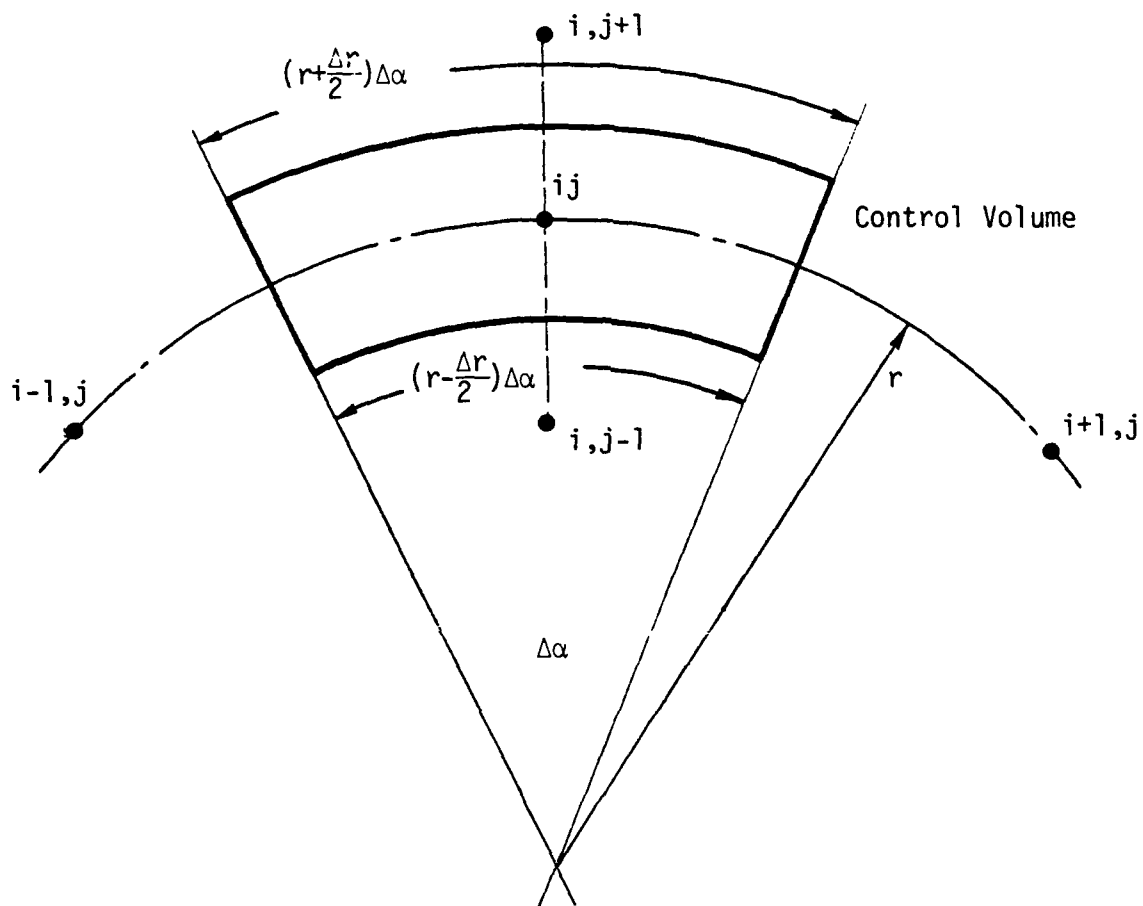


Figure 8-7. Control Volume in Cylindrical Coordinates.

$$\begin{aligned}
& + \frac{\beta \rho_c (r - \frac{\Delta r}{2}) \Delta \alpha}{12 \mu \Delta r} \left(\frac{h_{i \cdot j \cdot t} + h_{i \cdot j - 1 \cdot t}}{2} \right)^3 g_{i \cdot j \cdot t} \\
& + \frac{r \omega \Delta r}{2} h_{i \cdot j \cdot t} (1 - g_{i \cdot j \cdot t}) \\
& + r \Delta \theta \Delta r (1 - g_{i \cdot j \cdot t}) \frac{h_{i \cdot j \cdot t}}{\Delta t} \Big] \\
= & \frac{\beta \rho_c \Delta r}{12 \mu r \Delta \alpha} \left(\frac{h_{i \cdot j \cdot t} + h_{i+1 \cdot j \cdot t}}{2} \right)^3 (g_{i \cdot j \cdot t} - g_{i+1 \cdot j \cdot t}) \\
& + \frac{\beta \rho_c \Delta r}{12 \mu r \Delta \alpha} \left(\frac{h_{i \cdot j \cdot t} + h_{i-1 \cdot j \cdot t}}{2} \right)^3 (g_{i \cdot j \cdot t} - g_{i-1 \cdot j \cdot t}) \\
& + \frac{\beta \rho_c (r + \frac{\Delta r}{2}) \Delta \alpha}{12 \mu \Delta r} \left(\frac{h_{i \cdot j \cdot t} + h_{i \cdot j+1 \cdot t}}{2} \right)^3 (g_{i \cdot j \cdot t} - g_{i \cdot j+1 \cdot t}) \\
& + \frac{\beta \rho_c (r - \frac{\Delta r}{2}) \Delta \alpha}{12 \mu \Delta r} \left(\frac{h_{i \cdot j \cdot t} + h_{i \cdot j-1 \cdot t}}{2} \right)^3 (g_{i \cdot j \cdot t} - g_{i \cdot j-1 \cdot t}) \\
& + \frac{r \omega \Delta r}{2} (g_{i-1 \cdot j \cdot t} h_{i-1 \cdot j \cdot t} - g_{i \cdot j \cdot t} h_{i \cdot j \cdot t}) \\
& - r \Delta \theta \Delta r \theta_{i \cdot j \cdot t - \Delta t} (1 - g_{i \cdot j \cdot t - \Delta t}) \frac{h_{i \cdot j \cdot t - \Delta t}}{\Delta t} \\
& + \frac{r \omega \Delta r}{2} \left[\frac{1}{2} g_{i \cdot j \cdot t} g_{i-1 \cdot j \cdot t} (h_{i \cdot j \cdot t} - h_{i-1 \cdot j \cdot t}) \right. \\
& \quad \left. - \frac{1}{2} g_{i \cdot j \cdot t} g_{i+1 \cdot j \cdot t} (h_{i+1 \cdot j \cdot t} - h_{i \cdot j \cdot t}) \right] . \tag{8-20}
\end{aligned}$$

The film thickness for the squeeze wavy seal is described in detail in Reference [6], and is given as

$$h = h_o + w(r) + [v_o + (r - r_c) \phi_o] \cos(-n(\theta - \omega t)) , \quad (8-21)$$

where

$$w(r) = -\min\{[v_o + (r - r_c) \phi_o] \cos(-n(\theta - \omega t))\}$$

h_o = minimum film thickness

v_o = amplitude

ϕ_o = the maximum tilt of the seal

ω = angular speed

r_c = radius to centroid of seal ring

r = radial coordinate

α = angular coordinate

The boundary conditions are at

$$r = r_i , \quad p = p_i ,$$

$$r = r_o , \quad p = p_o , \quad (8-22)$$

where

r_i = inside radius of circular ring

r_o = outside radius of circular ring

p_i = seal inside pressure

p_o = seal outside pressure

The numerical data used here are

$$h_o = 0.246 \times 10^{-4} \text{ in.}$$

$$v_o = 0.72 \times 10^{-5} \text{ in.}$$

$$\phi_o = 0.23 \times 10^{-3}$$

$$\omega = 1800 \text{ rpm}$$

$$r_c = 1.9361 \text{ in.}$$

$$r_i = 1.90 \text{ in.}$$

$$r_o = 2.0875 \text{ in.}$$

$$p_i = 0$$

$$p_o = 500 \text{ psi}$$

$$\mu = 0.99 \times 10^{-7} \text{ lb}\cdot\text{s/in.}^2$$

$$\beta = 5000$$

The solution procedures for both conventional and squeeze wavy seals of 2-D case are similar to that of 1-D case. Again, it is shown that by reducing the size of the mesh, the value of the load support for a squeeze wavy seal approaches that of a standard wavy seal (Figure 8-8).

Deformable Body Squeeze Sliding Bearings/Seals

In the previous section, distortion caused by the pressure forces induced in a fluid film flow was not considered. In actual situations, the distortion does affect the lubrication performance. For example, according to Brighton, Hooke, and O'Donoghue [56], in a journal bearing, the effects of distortion include

- (1) reducing the peak pressure for a given load
- (2) increasing the eccentricity ratio for a given load
- (3) moving the location of minimum film thickness towards the cavitation zone
- (4) increasing the cavitation angle

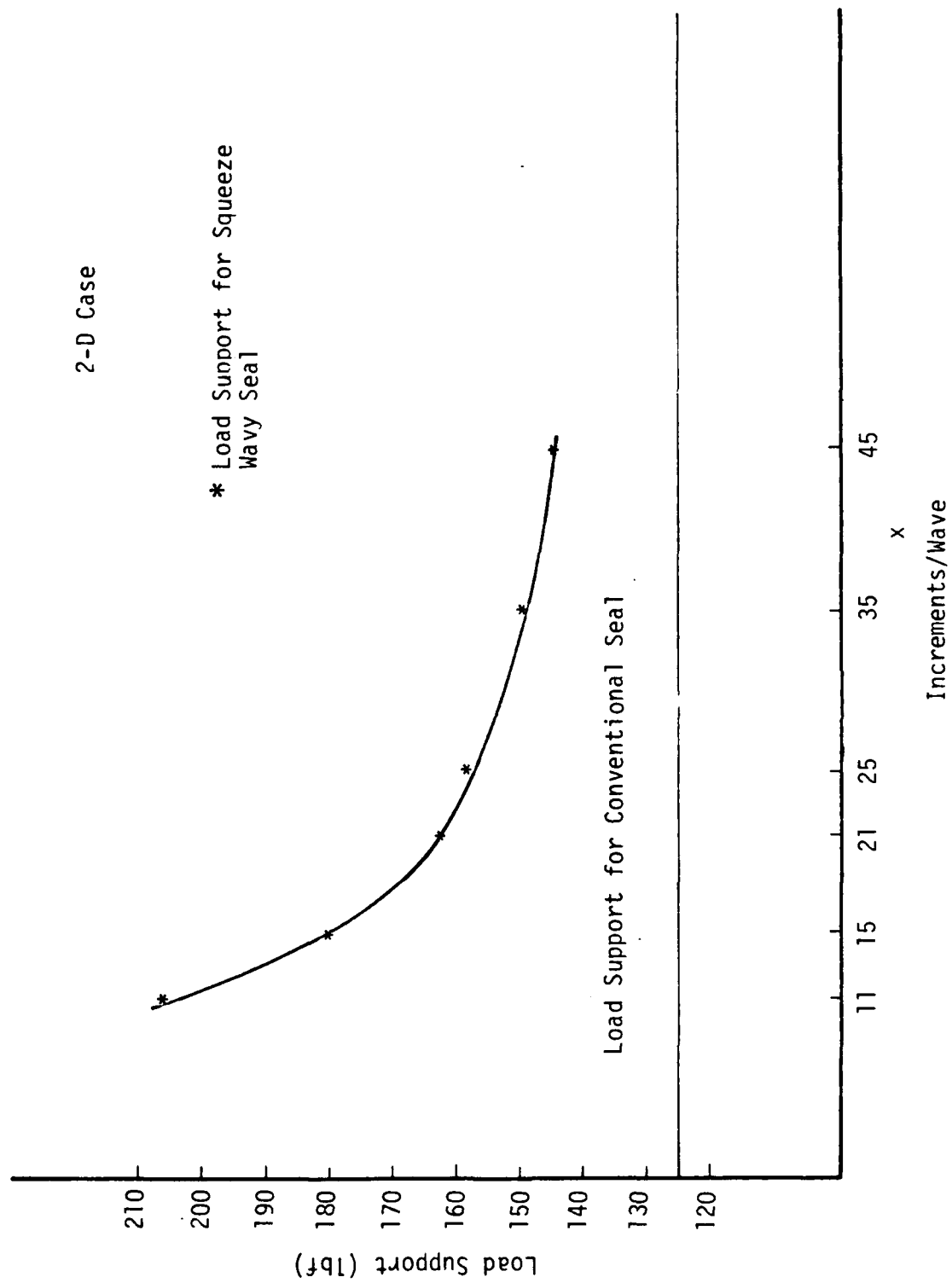


Figure 8-8. Effect of Mesh Size--Two Dimensional Case.

While it has been shown that the performance of rigid body squeeze wavy seals are identical to that of rigid body standard wavy seals, it would be interesting to find out about the performance of a deformable squeeze wavy seal/bearing which could be obtained by having a housing with variable stiffness along its length (as described earlier in this chapter, see Figure 8-3. In fact a new type of journal bearing (Figure 8-9) can be designed according to this concept.

In (Figure 8-9), it is shown that instead of having a circular shaft as that of classical journal bearings, a shaft constructed by n wavy surfaces (only three are shown in Figure 8-9) is placed and eccentrically rotated inside the bearing. The bearing is surrounded by an elastic housing having variable stiffness.

The bearing may be thought of as an infinite plate resting on a so-called Winkler foundation [57]. The foundation can be modeled as a series of closely spaced springs, each of which can deflect independently of its neighbors.

For simplicity, it is assumed that the length of the bearing is long as compared to its diameter and thus enabling us to first consider this problem as a one-dimensional case.

In Figure 8-10 it is shown that the bearing is represented by an infinite plate resting on an elastic foundation having variable stiffness, while the shaft with wavy surfaces is made to move at a velocity U_2 .

To find the solution of this problem, the deflection caused by pressure force must be first found.

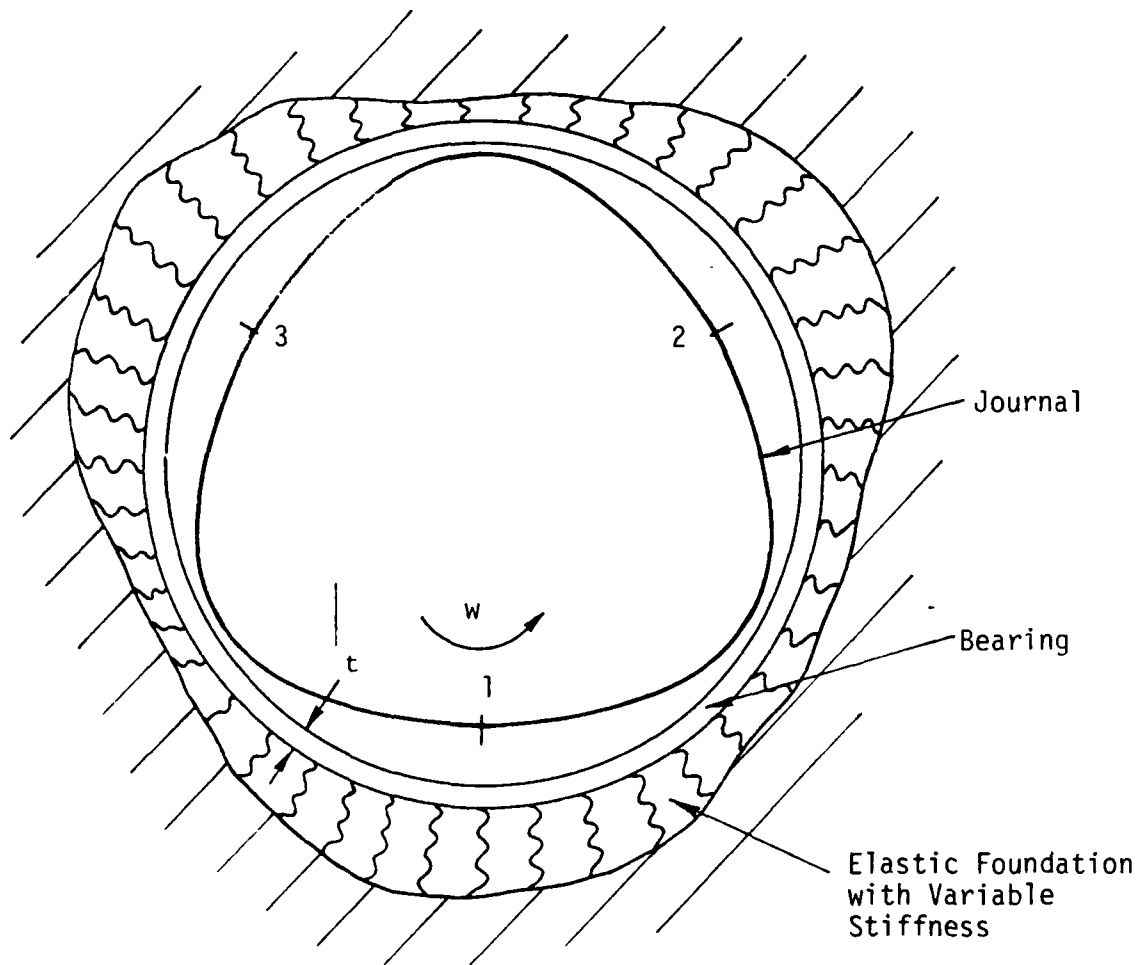


Figure 8-9. Wavy-Squeeze-Deformable Journal Bearing.

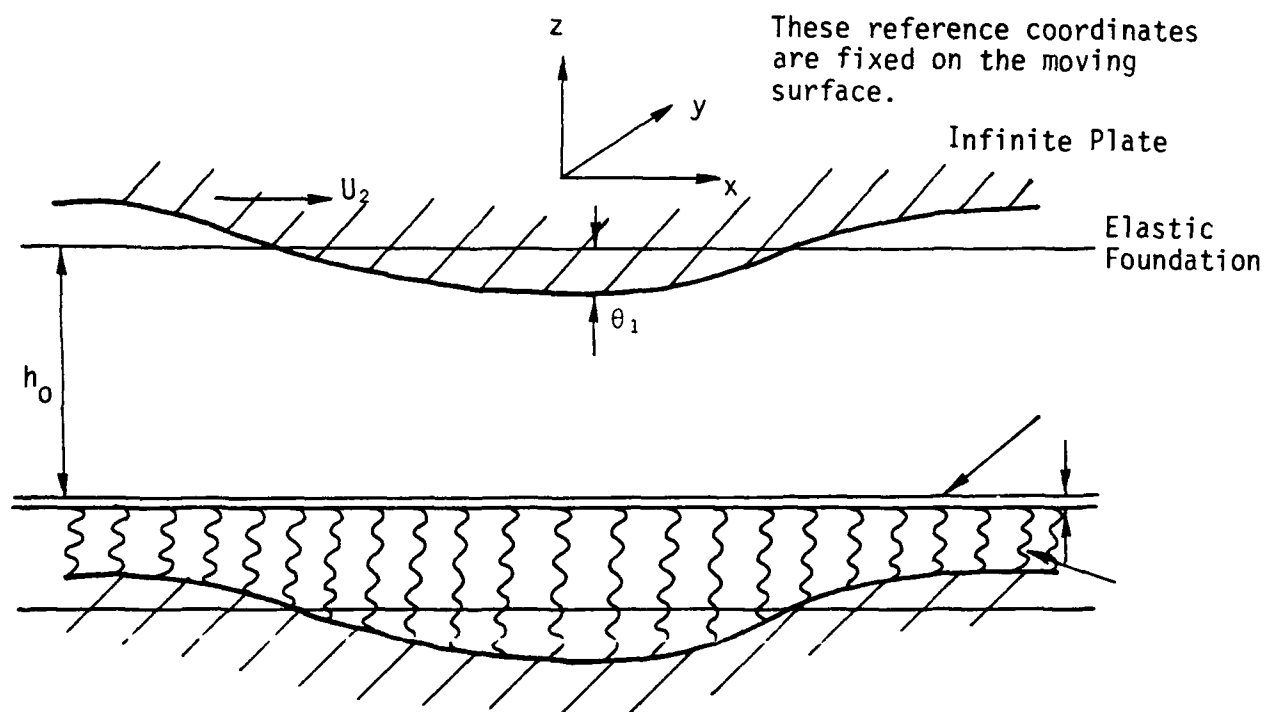


Figure 8-10. Wavy-Squeeze-Deformable Problem.

Deflection of An Infinite Plate Resting on an Elastic Foundation

Consider an infinitely long rectangular plate with uniform thickness t subjected to a uniformly distributed pressure force along its center line (Figure 8-11). The governing deflection equation for this plate can be represented by

$$D \left(\frac{\partial^4 W}{\partial x^4} + \frac{\partial^4 W}{\partial x^2 \partial y^2} + \frac{\partial^4 W}{\partial y^4} \right) = D - kW, \quad (8-23)$$

where

$$D = \frac{Et^3}{12(1 - \mu^2)}$$

W = deflection

k = foundation modulus

p = given pressure force

L = axial length of the bearing

If the axial length of the bearing is long enough, the variation of deflection along the y direction can be ignored, i.e., $W = W(x)$. In such a case, Equation (8-23) becomes

$$D \frac{d^4 W}{dx^4} = P - kW. \quad (8-24)$$

The value of k is assumed as

$$k(x) = C_1 + C_2 \sin(nx), \quad (8-25)$$

where

C_1, C_2 = constant

n = number of wave.

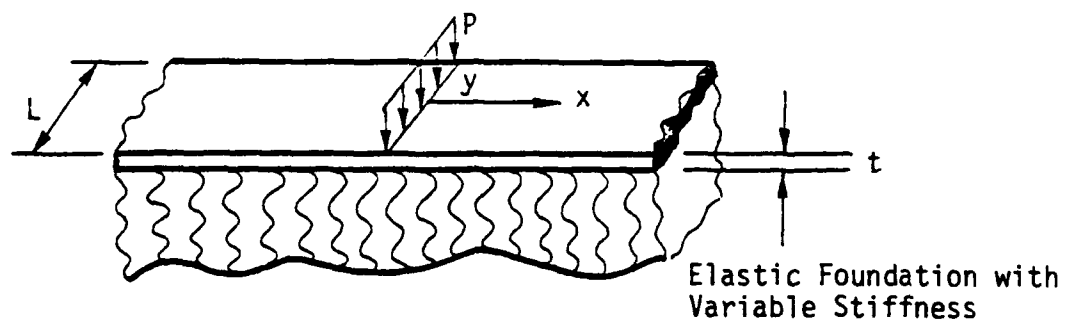


Figure 8-11. Plastic Foundation.

The reason why k is chosen as a sine wave is based on the fact that the original undeformed film thickness is chosen as a cosine wave. Under this condition, the effects of this variable stiffness can be clearly expressed.

The boundary conditions to be used here with Equation (8-24) are

$$x \rightarrow \infty \quad W = 0 , \quad (8-26)$$

$$M = D \frac{d^2 W}{dx^2} = 0 , \quad (8-27)$$

$$x \rightarrow -\infty \quad W = 0 , \quad (8-28)$$

$$M = D \frac{d^2 W}{dx^2} = 0 , \quad (8-29)$$

where M is moment.

An approximate solution can be obtained by changing Equation (8-20) into a finite difference equation, using the above boundary conditions, and solving with Gauss elimination method.

Since the value of k is changed from point to point along this infinite plate, it is necessary to find the deflection curves caused by a unit pressure force acting on each location.

By using the data for these deflection curves and the superposition method, one can easily obtain a deflection curve caused by any kind of pressure distribution.

Governing Equation of the Squeeze Wavy Bearing with Variable Stiffness

The assumptions upon which the present solution is based are similar to those discussed earlier except that elastic distortion is

The film thickness is now given as

$$h = h_i + \delta , \quad (8-30)$$

where

h_i = original film thickness under no deformation condition

δ = deformation caused by pressure force

Here h_i is expressed as (see Figure 8-10)

$$h_i = h_o + \epsilon_1 \cos(-n(Ut - x)) . \quad (8-31)$$

To solve Equation (8-16), the boundary conditions are the same as that described previously, that is, pressure is zero at the point of maximum film thickness and cavitation occurs where it needs to. To solve for a one-dimensional undeformed squeeze wavy seal, the solution procedure is described as follows:

- (1) At each time instant the pressure distribution under which no deformation is considered is first obtained by using over-relaxation method (relaxation factor = 1.7).
- (2) This pressure distribution is then used to determine the distortion. The distortion alters the film shape and consequently a new pressure distribution is developed. This iteration process is continued until a final solution which satisfies both the hydrodynamic and elastic deformation requirements of the system is reached.
- (3) The pressure distribution and film thickness obtained from the present time instant are substituted into Equation

(8-16). A new pressure distribution for the next time instant is generated.

For all the iteration processes, an effective way to speed up the convergence is by adjusting the pressure distribution at each iteration. In other words, instead of only using the pressure distribution obtained by the previous iteration, a weighted mean of those pressures obtained from the previous iteration is also applied to the next iteration, i.e.,

$$p(j) = p(j - 1) * CONV + (1 - CONV) * p_m , \quad (8-32)$$

where

j = iteration number

CONV = convergent factor (0.58 is used)

p_m = weighted mean of pressure

Numerical Results

The numerical data used here are

$$h_o = 4.5 \times 10^{-4} \text{ in.}$$

$$\epsilon_1 = 5.0 \times 10^{-5} \text{ in.}$$

$$n = 3$$

$$E = 3 \times 10^6 \text{ lb/in.}^2$$

$$\mu = 0.99 \times 10^{-7} \text{ Reyn.}$$

$$\nu = 0.3$$

$$\omega = 1800 \text{ rpm}$$

$$k = 3 \times 10^6 + 1 \times 10^6 \sin nx \text{ lb/in.}$$

$$\beta = 500$$

$$\Delta t = 0.3175 \times 10^{-3} \text{ sec}$$

The numerical result for the present problem is shown in Figure 8-12. Since the reference coordinates are fixed on the shaft (or moving surface), the stiffness curves of the elastic foundation are different for each time instant. In order to compare the load support to a conventional bearing, a minimum film thickness is held as constant all the time ($h_{\min} = 4.3 \times 10^{-4}$ in.). It is shown that the values of the load support vary within the range of 82.7 ~ 120 lb. Maximum load support occurs at time $t = (28 - 1) * \Delta t$ and minimum load support occurs at time $t = (11 - 1) * \Delta t$. The stiffness curves for both maximum and minimum load support are shown in Figure 8-13. It is found that minimum load support occurs at location where the stiffness curves has the same shape as that of the film thickness. The maximum load occurs at a location where the stiffness curve is just opposite to the shape of the film thickness. Since the deformation is small as compared to the original undeformed film thickness, the shape of the film thickness can be roughly described as Figure 8-13(c).

Conclusion

Due to the convergence difficulty encountered in the iteration process, only small elastic deformation (at the order of 10^{-6} in.) is considered here. The difference in load support found so far is small. Further investigation is still necessary to fully understand the performance of the squeeze wavy seal/bearing.

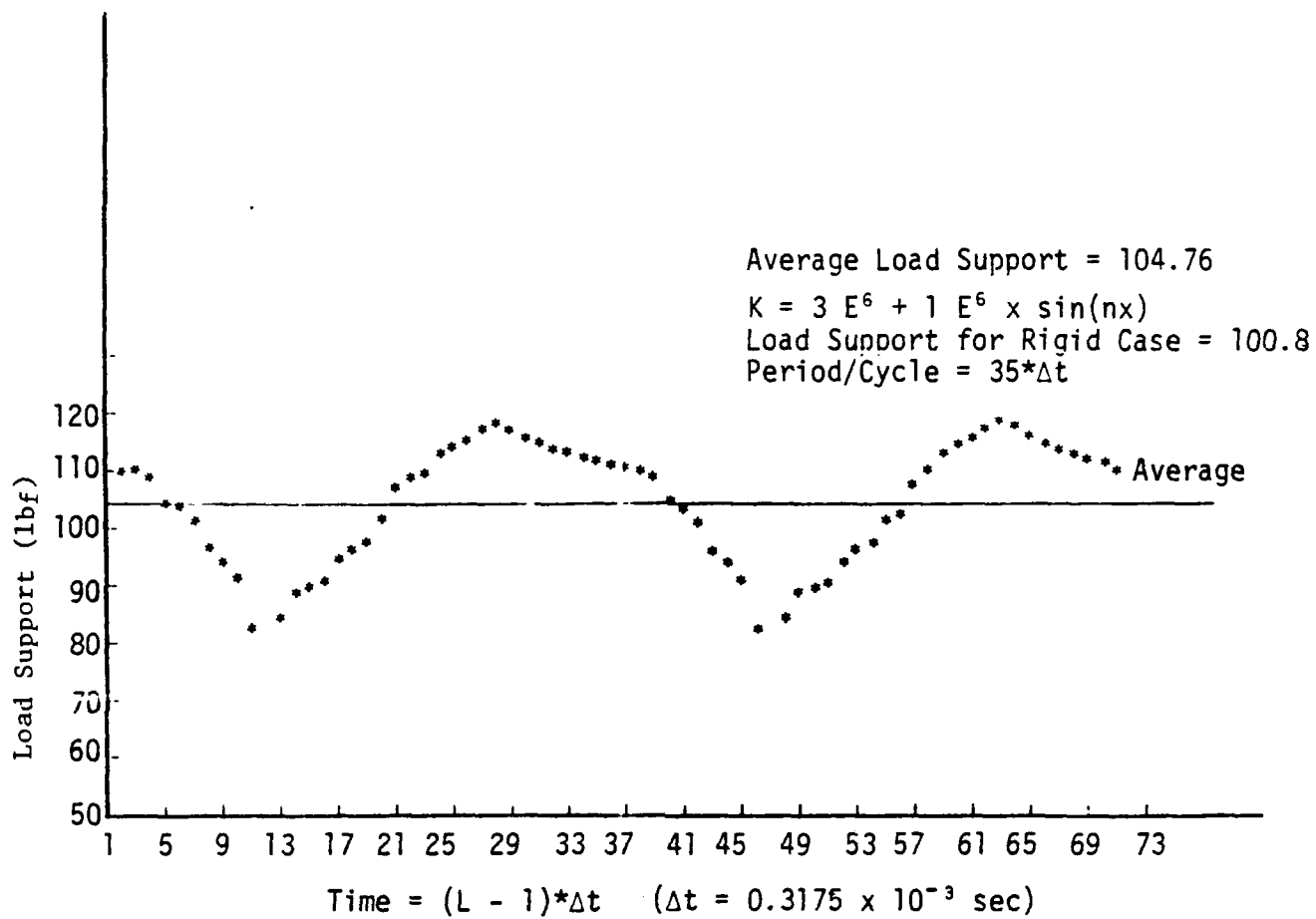
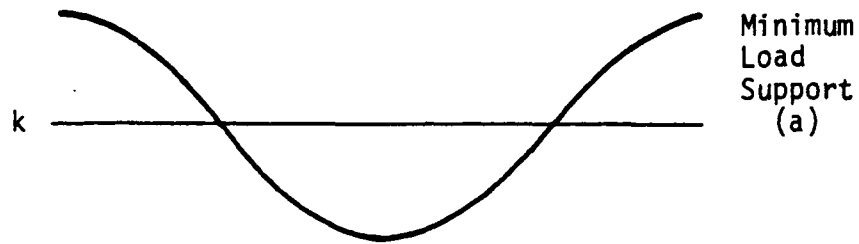
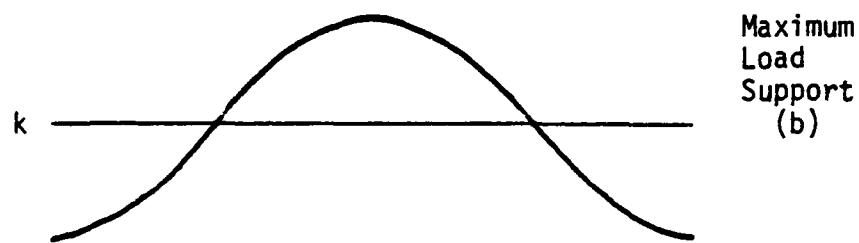


Figure 8-12. Wavy-Squeeze-Deformable Seal Load Support.

At Time $L = 11$



At Time $L = 28$



Film Thickness

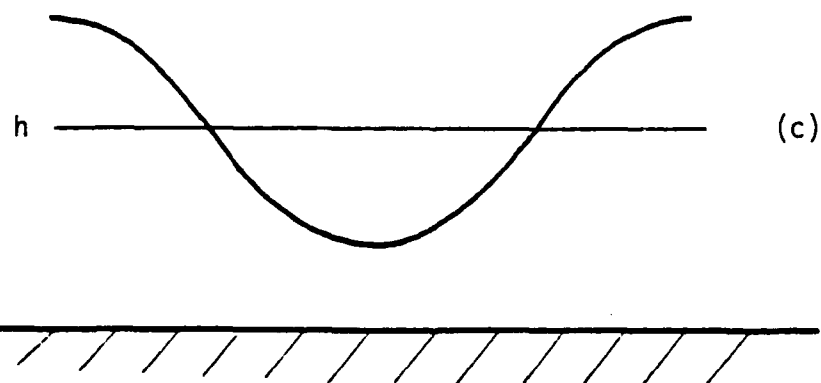


Figure 8-13. Condition of Maximum and Minimum Load Support.

CHAPTER 9

SUMMARY AND CONCLUSIONS

Nine Wave Seal--Design 1

After initial design problems were overcome, this seal ran for the 2000 hour test successfully. Face wear was very low (530 μ in.) and would be expected to provide more than adequate wear life for 150000 hours of operation. Torque was somewhat higher and leakage somewhat lower than predicted indicating that not as much wave as desired was actually present on the seal faces.

It was concluded that even though this seal design operated successfully, it is too complex to manufacture for practical applications because of the many small flow passages and machined pads. A simpler type of design was sought.

Nine Wave Seal--Design 2

A much simpler and easier to fabricate design was created. This design uses an all carbon seal ring with no cut pads. Fifty-four O-ring seal pistons are used on the inside of the seal. This part is manifolded using soldered tubing. These parts were machined from brass and protected by being encapsulated in epoxy.

This seal design was fabricated and ran for 2000 hours with no significant problems. Wear was 355 μ in. average, lower than for design one (0.026 in. wear in 150000 hours). Torque was slightly higher and leakage lower than predicted.

Even though this design was successful and relatively simple to fabricate, it was decided that the uncertainty in reliability brought about by the pressure pads and O-ring system plus the complexity of the essential auxiliary pressure system were still significant disadvantages to the practical application of the wavy seal. Thus, an even simpler design and waviness drive system were sought.

Nine Wave Seal--Design 3

After many different approaches were considered and discarded, it was decided that placing the unique wave shape permanently on the hard face produced, a wavy seal with the greatest possible advantage in that no auxiliary support or pressure system is needed at all. The design has the same low leakage--low torque characteristics as the previous design. There is only one potential limitation and that is that the high spots may wear off of the hard face. Unlike the previous designs, both faces are not wiped everywhere. There are many reasons to believe that the wave will not wear off during the practical lifetime of the seal, but only testing will tell for sure. Short term tests run to date look very promising.

This idea represents a compromise between having the ultimate in wearing faces with the uncertainty of auxiliary systems and having possibly a somewhat shorter lifetime with near absolute reliability. Given the excellent wear resistance of SiC, there is a high probability that the hard face will perform adequately.

A new type of grinding fixture was designed to grind the nine wave tilted face. The fixture is based on a flexure mount so that it can be made of only moderately precision parts.

500 Hour Parallel Face Test

A 500 hour variable condition test was run using the flat faced version of the first nine wave seal. The wear on the seal was surprisingly low and torque was very high (many times exceeding the capacity of the test machine). It was concluded that while the wear is low the high friction of this type of seal will make long term operation somewhat uncertain because of possible material damage due to high interface temperature.

Force Transducer Design

A new type of integral force transducer was designed in an attempt to find a better force applier than the O-ring systems. This system was based on using a porous medium and an epoxy diaphragm. Several experiments were run. The device worked but was judged too unreliable for the intended purpose.

Torsion of Composite Section

Some unique work was performed to develop a method to evaluate the torsional stiffness of cross sections made up of two materials.

Warping Evaluation

For nine wave seal design 2, warp again (as in design 1) turned out to add significantly to the stiffness in the torsional deformation required to produce the wave. Three-dimensional finite element modeling was performed to evaluate the warping function. In this design, measured waviness is reasonably close to predicted waviness in a large part because warping was properly accounted for.

Two Ring Contact Model

The two ring contact model was analyzed, checked experimentally, and proved to be very useful in evaluating stiffness effects in actual submarine seals. In particular, studies show

- 1) Submarine seal leakage is readily caused by lock ring groove out of flatness.
- 2) Submarine seals cannot close up gaps of the size found in typical leakers--they are very stiff.
- 3) The model successfully predicted the effects placing a shim in the lock ring groove.
- 4) The variation of stiffness at the seal joint in conjunction with the large moment in the ring causes the seal to just separate.
- 5) Even a highly compliant seal will not close off an opening caused by a 0.001 in. segment shift.

Magnetic Seals

The two ring contact model was used to predict leakage caused by waviness in small magnetic seals caused by waviness.

Nonlinear Joint Model

It became clear in using the two ring contact model that a better model of the seal bolted joint is needed. Some preliminary FEM modeling of the joint has been made. More extensive modeling and experiments are planned.

Seal Conformability

Various stiffness studies and the design of the wavy seal show how tangential stiffness must be carefully considered in seal design. In the case of the wavy seal, nine waves are needed so that they do not flatten out and cause the wavy effect to be diminished. In the case of submarine seals and the magnetic seals evaluated herein, they are too stiff to properly close to minimize leakage. With the various stiffness models tools are now in hand to carefully make these evaluations.

Friction and Wear Tester

A new hydrostatic bearing support was designed and fabricated for the friction and wear test machine. The design reduced the influence of friction on measurements to zero and allows accurate application of the load. The bearing is designed to operate on a 0.00075 in. film of water.

Microasperity Lubrication

Using the above apparatus extensive measurements were made of carbon-WC friction in water independent of hydrostatic and conventional hydrodynamic effects. It was found that friction is greatly reduced with increasing speed suggesting hydrodynamic effects. By varying ambient water pressure, experiments were run to reduce the effectiveness of microasperity cavitation--the hypothesized mechanism. There was no effect. Thus it appears that microasperity lubrication is not causing this beneficial lubrication. Given the importance of this unknown mechanism (in that it might be even more effectively and widely applied if understood), more work is being conducted to begin to understand this strong speed effect.

Seal Analysis

Several automated computer programs have been developed for the purpose of expediting design and analysis of seals. The first program creates a refined finite difference mesh for a completely arbitrary cross section and calculates the circular ring section properties and the torsional stiffness by solving the Poisson equation. Input is very simple and the program does all of the work automatically.

A second program creates a mesh for FEM input automatically. A third program solves the heat transfer equations to find the temperature distributions in a two ring seal (with interface) and the thermal rotation. Arbitrary convective boundaries can be easily input. The analysis is completely automatic and serves as a very powerful design tool to look at the effect of thermal rotation on design.

Squeeze Seal/Bearings

It has been shown that certain squeeze seal concepts are hydrodynamically identical to the present wavy seal but may be simpler to implement.

A new type of wavy squeeze/seal bearing which uses an elastic foundation of variable stiffness is being evaluated.

General

A great deal has been learned and written about in terms of the proper design considerations for wavy seals as well as conventional seals. The role of stiffness, deflection, hydrodynamic and hydrostatic lubrication and thermal rotation have been much better understood. Many mechanic's problems have been solved to obtain this understanding.

Yet there remain uncertainties in seal design and related lubrication problems where more attention would be most worthwhile. In particular a better understanding of the fundamental lubrication mechanisms in carbon-hard face sliding system is sorely needed. Identification of the existence and content of cavitation in water hydrodynamics has never been done. Work on the role of microasperities in lubrication remains unfinished. Finally obtaining an understanding of a whole new type of bearing/seal--the squeeze seal/bearing has just begun. There remain many challenges in these areas of tribology.

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APPENDIX A

FINITE DIFFERENCE SOLUTION
TO THE TORSION PROBLEM OF A
COMPOSITE CROSS SECTION

Considering the rectangular cross section as shown in Figure A1 where the boundaries align with the x and y axis, the free boundary condition from Chapter 3 is

$$\left(\frac{\partial \psi}{\partial x} - y\right) \frac{dy}{ds} - \left(\frac{\partial \psi}{\partial y} + x\right) \frac{dx}{ds} = 0 \quad (1)$$

For corner #1, $\frac{dy}{ds} = 1$ and $\frac{dx}{ds} = 1$. So the finite difference formulation

gives

$$\psi_{(I,J)} = \left[\frac{(x(I+1) - x(I)) (y(J+1) - y(J))}{(x(I+1) - x(I)) + (y(J+1) - y(J))} \right] x$$

$$\left[\frac{\psi_{(I+1,J)}}{x(I+1) - x(I)} + \frac{\psi_{(I,J+1)}}{y(J+1) - y(J)} - y(J) - x(I) \right] \quad (2)$$

For the boundary between corners #1 and #2, we have that $\frac{dy}{ds} = 1$ and

$\frac{dx}{ds} = 0$, so this gives

$$\psi_{(I,J)} = \psi_{(I+1,J)} - (x(I+1) - x(J)) y(J) \quad (3)$$

Corner #2 has $\frac{dy}{ds} = 1$ and $\frac{dx}{ds} = 1$ so

$$\psi_{(I,J)} = \left[\frac{(x(I+1) - x(I)) (y(J) - y(J-1))}{(x(I+1) - x(I)) (y(J) - y(J-1))} \right] x$$

$$\left[\frac{\psi_{(I+1,J)}}{x(I+1) - x(I)} + \frac{\psi_{(I,J-1)}}{y(J) - y(J-1)} - y(J) - x(I) \right] \quad (4)$$

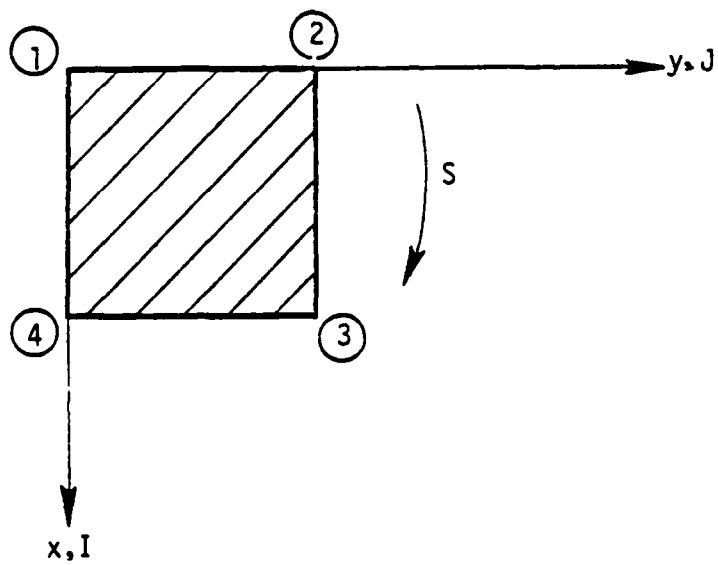


Figure A1. Simple Cross Section for Boundary Derivations.

For the boundary between corners 2 and #3 we have $\frac{dy}{ds} = 0$ and $\frac{dx}{ds} = 1$; so

$$\psi_{(I,J)} = \psi_{(I,J-1)} - (y(J) - y(J-1)) x(I) \quad (5)$$

Corner #3 has $\frac{dy}{ds} = -1$ and $\frac{dx}{ds} = 1$ which gives

$$\psi_{(I,J)} = \left[\frac{(x(I) - x(I-1)) (y(J) - y(J-1))}{(x(I) - x(I-1)) + (y(J) - y(J-1))} \right] x$$

$$\left[\frac{\psi_{(I-1,J)}}{x(I) - x(I-1)} + \frac{\psi_{(I,J-1)}}{y(J) - y(J-1)} + y(J) - x(I) \right] \quad (6)$$

The boundary 3 - 4 has $\frac{dy}{ds} = -1$ and $\frac{dx}{ds} = 0$; so

$$\psi_{(I,J)} = \psi_{(I-1,J)} + (x(I) - x(I-1)) y(J) \quad (7)$$

For corner #4, $\frac{dy}{ds} = -1$ and $\frac{dx}{ds} = -1$ which gives

$$\psi_{(I,J)} = \left[\frac{(x(I) - x(I-1)) (y(J+1) - y(J))}{(x(I) - x(I-1)) + (y(J+1) - y(J))} \right] x$$

$$\left[\frac{\psi_{(I-1,J)}}{x(I) - x(I-1)} + \frac{\psi_{(I,J+1)}}{y(J+1) - y(J)} + y(J) + x(I) \right] \quad (8)$$

And finally for the boundary 4 - 1, we have $\frac{dy}{ds} = 0$ and $\frac{dx}{ds} = -1$

which gives

$$\psi_{(I,J)} = \psi_{(I,J+1)} + (y(J+1) - y(J)) x(I) \quad (9)$$

For all internal nodes we have

$$\begin{aligned} \psi_{(I,J)} = & AIM1(I,J)\psi_{(I-1,J)} + AIP1(I,J)\psi_{(I+1,J)} + AJM1(I,J)\psi_{(I,J-1)} \\ & + AJP1(I,J)\psi_{(I,J+1)} \end{aligned} \quad (10)$$

where

$$\begin{aligned} AIJ &= \frac{-2}{(x(I-1) - x(I))(x(I+1) - x(I))} \\ &\quad - \frac{2}{(y(J-1) - y(J))(y(J+1) - y(J))} \\ AIM1(I,J) &= \frac{2}{(AIJ)(x(I-1) - x(I))(x(I-1) - x(I+1))} \\ AIP1(I,J) &= \frac{-2}{(AIJ)(x(I+1) - x(I))(x(I-1) - x(I+1))} \\ AJM1(I,J) &= \frac{2}{(AIJ)(y(J-1) - y(J))(y(J-1) - y(J+1))} \\ AJP1(I,J) &= \frac{-2}{(AIJ)(y(J+1) - y(J))(y(J-1) - y(J+1))} \end{aligned} \quad (11)$$

Now, taking the analysis one step further and considering a cross section of two different materials as shown in Figure A2, all boundary conditions are the same as previously derived. The only additional boundary condition is that across 2 - 5. For this boundary

$\frac{dy}{ds} = 0$ and $\frac{dx}{ds} = 1$ and $\psi_1 = \psi_2 = \psi$ at the boundary. This gives

$$G_1 \left(\frac{\partial \psi}{\partial y} + x \right) = G_2 \left(\frac{\partial \psi}{\partial y} + x \right) \quad (12)$$

Finite differencing gives

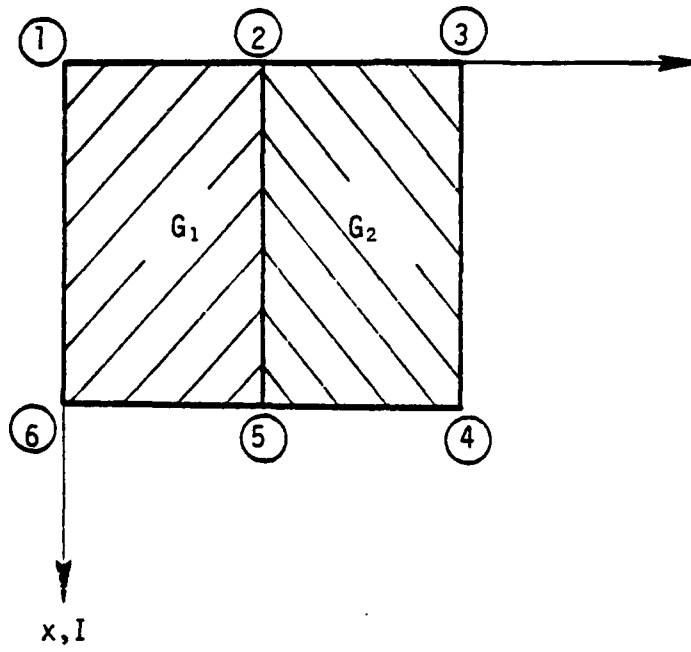


Figure A2. Composite Cross Section.

$$\psi_{(I,J)} = \frac{1}{1 + A(J)} \left[A(J) \left(\psi_{(I,J+1)} + (y(J+1) - y(J))x(I) \right) + \psi_{(I,J-1)} - (y(J) - y(J-1))x(I) \right] \quad (13)$$

where

$$A(J) = \frac{G_2}{G_1} \frac{y(J) - y(J-1)}{y(J+1) - y(J)} \quad (14)$$

A computer program was written using the derived boundary conditions. Results for the cross section of Figure A2 are given in Table A1. The results for G^*J_0 are those for the composite cross section method of solution and for the method not considering a composite section, i.e., the section has only one G value. These runs were made to check the consistency of G^*J_0 when the type of mesh and size were changed. The results show that for a non-uniform mesh in both x and y, the mesh must be refined. For the 29 x 29 matrix, there is only an 8% difference between a uniform and non-uniform mesh. However, for a 9 x 9 mesh, the error using a non-uniform mesh is large.

TABLE A-1

Run #	Modulus of Rigidity	Type Mesh	Size of Mesh	G* J _θ	
				w/ Composite Crossection	w/o Composite Crossection
1	$G_1 = 11.5 \times 10^6$	Uniform in both x and y	9 x 9	2427	2460
	$G_2 = 11.5 \times 10^6$				
2	$G_1 = 11.5 \times 10^6$	Uniform in both x and y	9 x 9	811	2460
	$G_2 = 1.25 \times 10^6$				
3	$G_1 = 11.5 \times 10^6$	Non-uniform in y only	9 x 9	588	2416
	$G_2 = 1.25 \times 10^6$				
4	$G_1 = 11.5 \times 10^6$	Non-uniform in y only	9 x 9	2145	2416
	$G_2 = 11.5 \times 10^6$				
5	$G_1 = 11.5 \times 10^6$	Uniform in both x and y	29 x 29	2567	2566
	$G_2 = 11.5 \times 10^6$				
6	$G_1 = 11.5 \times 10^6$	Uniform in both x and y	29 x 29	866*	2566
	$G_2 = 1.25 \times 10^6$				
7	$G_1 = 11.5 \times 10^6$	Non-uniform in y only	29 x 29	820	2560
	$G_2 = 1.25 \times 10^6$				
8	$G_1 = 11.5 \times 10^6$	Non-uniform in both x and y	29 x 29	800	2553
	$G_2 = 1.25 \times 10^6$				

*Reference Solution.

11/2/82

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READY
LIS
10 C      STIFFNESS CONSTANT (G*J0) CALCULATED USING A FINITE DIFFERENCE
20 C      METHOD AND CONSIDERING A COMPOSITE CROSSSECTION
25 C
30          DIMENSION X(40), Y(40), AIM1(40,40), AIP1(40,40), AJM1(40,40)
40          DIMENSION AJPL(40,40), BPHS(40,40), P(40,40), PSI(40,40), R(40)
50          ERR=1.E-06
55          G1=11.5E+06
56          G2=1.05E+06
60          BB1=.08
70          BB2=.1
80          BB3=.2
90          AA1=.08
100         AA2=.1
105         AA3=.15
106         AA4=.2
110         I1=8
120         I2=15
130         I3=29
140         J1=8
150         J2=15
155         J3=22
156         J4=29
160         XX=0.0
170         X(1)=XX
180         DX=BB1/(I1-1)
190         DO 10 I=2, I3
200             IF (I.GT.I1) DX=(BB2-BB1)/(I2-I1)
210             IF (I.GT.I2) DX=(BB3-BB2)/(I3-I2)
220             XX=XX+DX
230             X(I)=XX
240     10 CONTINUE
250         YY=0.0
260         Y(1)=YY
270         DY=AA1/(J1-1)
280         DO 20 J=2, J4
290             IF (J.GT.J1) DY=(AA2-AA1)/(J2-J1)
295             IF (J.GT.J2) DY=(AA3-AA2)/(J3-J2)
296             IF (J.GT.J3) DY=(AA4-AA3)/(J4-J3)
300             YY=YY+DY
310             Y(J)=YY
320     20 CONTINUE
330         IMAX=I3-1
340         JMAX=J4-1
350         DO 30 I=2, IMAX
360             DO 30 J=2, JMAX
370                 AIG=-2./X(I-1)-X(I)+X(I+1)+Y(J)
380                 AID=AIG-2./Y(J-1)-Y(J)+Y(J+1)+X(I)
390                 AIM1(I,J)=2./X(I-1)-X(I)+X(I+1)+Y(J-1)+Y(J+1)/AIG
400                 AIP1(I,J)=-2./X(I-1)+X(I)-X(I+1)+Y(J-1)+Y(J+1)/AID
410                 AJM1(I,J)=2./Y(J-1)-Y(J)+Y(J+1)+X(I-1)+X(I+1)/AIG
420                 AJPL(I,J)=-2./Y(J-1)+Y(J)-Y(J+1)+X(I-1)+X(I+1)/AIP
430                 BPHS(I,J)=2./AIG
440     30 CONTINUE
450         DO 40 I=1, I3
460             DO 40 J=1, J4
470                 P(I,J)=0.0
480     40 CONTINUE
490         ONEG=1.5
500         TWO=1.-ONEG

```

```

520      AMAX=0.0
530      DO 50 I=2,IMAX
540          JMAX=J4-1
550          JMIN=2
560 C      IF(I.GT.I1) JMAX=J1-1
570 C      IF(I.GT.I2) JMAX=J2-1
580      DO 50 J=JMIN,JMAX
590          POLD=P(I,J)
600          PNEW=AIM1(I,J)*P(I-1,J)+AIP1(I,J)*P(I+1,J)
610          PNEW=PNEW+AJM1(I,J)*P(I,J-1)+AJP1(I,J)*P(I,J+1)+BPMS(I,J)
620          P(I,J)=PNEW*OMEG+OMO*POLD
630          VAL=ABS(P(I,J)-POLD)
640          IF(VAL.GT.AMAX) AMAX=VAL
650      50 CONTINUE
660      IF(AMAX.LT.ERR) GO TO 80
670      50 CONTINUE
680      WRITE(6,70)
690      70 FORMAT(2X,'EXCEEDS ITERATIONS ')
700      80 JMAX=J4-1
710      IMAX=I3-1
720      AJ=0.0
730      DO 90 I=1,IMAX
740          DO 90 J=1,JMAX
750              Z=P(I,J)+P(I+1,J)+P(I+1,J+1)+P(I,J+1)
760              AJ=AJ+.25+2*(X(I+1)-X(I))*(Y(J+1)-Y(J))
770      90 CONTINUE
780      AJ=2.*AJ
790      AJO=AJ*G1
800 C
810 C      ADDITION TO AJO TO APPROXIMATE TOTAL STIFFNESS
820 C
830      WRITE(6,100) AJO,IT
840      100 FORMAT(5X,'G=00= ',E10.5,2X,'ITERATIONS=',I4,7X)
850      DO 105 I=1,I3
860          DO 105 J=1,J4
870              PSI(I,J)=0.0
880      105 CONTINUE
890      JMAX=J4
900      JMIN=(JMAX+1)/2
910      DO 120 I=1,I3
920          JMAX=J4
930          JMIN=1
940      DO 120 J=JMIN,JMAX
950          A(J)=(Y(J)-Y(J-1))/(Y(J+1)-Y(J))*G2/G1
960          PSIOld=PSI(I,J)
970          IF(I.EQ.1) GO TO 111
980          IF(I.EQ.I3) GO TO 114
990          IF(J.EQ.JMIN OR J.EQ.JMAX) GO TO 113
1000         IF(J.EQ.JMID) GO TO 110
1010         PSINEW=AIM1(I,J)*PSI(I-1,J)+AIP1(I,J)*PSI(I+1,J)
1020         PSINEW=PSINEW+AJM1(I,J)*PSI(I,J-1)+AJP1(I,J)*PSI(I,J+1)
1030         GO TO 119
1040      110 PSINEW=1./4.*A(J)*(A(J)*PSI(I,J+1)+(Y(J+1)
1050         -Y(J))*X(I)+PSI(I,J-1)+(Y(J)-Y(J-1))*X(I))
1060         GO TO 119
1070      111 IF(J.EQ.JMIN OR J.EQ.JMAX) GO TO 112
1080         IF(J.EQ.JMID) PSINEW=1./4.*A(J)*(A(J)*(PSI(I,J+1)+(Y(J+1)
1090         -Y(J))*X(I)+PSI(I,J-1)+(Y(J)-Y(J-1))*X(I))
1100         IF(J.NE.JMID) PSINEW=PSI(I+1,J)-(X(I+1)-X(I))*Y(J)
1110         GO TO 119
112         IF(J.EQ.JMIN) PSINEW=(X(I+1)-X(I))*(Y(J+1)-Y(J-1))
113         +X(I+1)*X(I)+(Y(J+1)-Y(J-1))*PSI(I+1,J)+(X(I+1)-X(I)
114         +PSI(I,J+1)*(Y(J+1)-Y(J-1))-Y(J)*X(I))

```

```

1090      IF(J.EQ.JMAX) PSINew=((X(I-1)-X(I))*Y(J)-Y(J-1))/
1091      *      ((X(I+1)-X(I))*(Y(J)-Y(J-1)))+(PSI(I+1,J)/(X(I+1)-X(I))
1092      *      +PSI(I,J-1)/(Y(J)-Y(J-1))-Y(J)-X(I))
1120      GO TO 119
1130      113 IF(J.EQ.JMIN) PSINew=PSI(I,J+1)+(Y(J+1)-Y(J))*X(I)
1140      IF(J.EQ.JMAX) PSINew=PSI(I,J-1)-(Y(J)-Y(J-1))*X(I)
1150      GO TO 119
1160      114 IF(J.EQ.JMIN.OR.J.EQ.JMAX) GO TO 115
1161      IF(J.EQ.JMID) PSINew=1./4*(A(J)+A(J+1)*(PSI(I,J+1)+(Y(J+1)
1162      *      -Y(J))*X(I))+PSI(I,J-1)-(Y(J)-Y(J-1))*X(I))
1170      IF(J.NE.JMID) PSINew=PSI(I-1,J)+(X(I)-X(I-1))*Y(J)
1180      GO TO 119
1190      115 IF(J.EQ.JMIN) PSINew=((X(I)-X(I-1))*(Y(J+1)-Y(J)))/
1200      *      ((X(I)-X(I-1))*(Y(J+1)-Y(J)))+(PSI(I-1,J)/(X(I)-X(I-1))
1210      *      +PSI(I,J+1)/(Y(J+1)-Y(J))+Y(J)+X(I))
1220      IF(J.EQ.JMAX) PSINew=((X(I)-X(I-1))*(Y(J)-Y(J-1)))/
1230      *      ((X(I)-X(I-1))*(Y(J)-Y(J-1)))+(PSI(I-1,J)/(X(I)-X(I-1))
1240      *      +PSI(I,J-1)/(Y(J)-Y(J-1))+Y(J)-X(I))
1250      GO TO 119
1260      119 PSI(I,J)=PSINew+OMEG+GMD+PSIOLD
1270      VAL=ABS(PSI(I,J)-PSIOLD)
1280      IF(VAL.GT.AMAX) AMAX=VAL
1290      120 CONTINUE
1300      121 IF(AMAX.LT.ERR) GO TO 140
1310      120 CONTINUE
1320      WRITE(6,70)
1330      140 CONTINUE
1335      DO 1000 I=1,13,7
1337      WRITE(6,2000) (PSI(I,J),J=1,J4,7)
1338      1000 CONTINUE
1339      2000 FORMAT(1X,5F8.5,/)
1340      C
1350      C      INTEGRATION ROUTINE
1360      C
1370      145 JMAX=J4-1
1380      IMAX=I4-1
1385      ATHETA=0.0
1390      DO 160 I=1,IMAX
1400      DO 160 J=1,JMAX
1410          DX=X(I+1)-X(I)
1420          DY=Y(J+1)-Y(J)
1425          IF(J.LT.JMID) G=G1
1426          IF(J.GE.JMID) G=G2
1430          DPSIDY=(PSI(I,J+1)-PSI(I,J)+PSI(I-1,J+1)-PSI(I+1,J))/
1440      *      /2./DY
1450          DPSIDX=(PSI(I+1,J)-PSI(I,J)+PSI(I+1,J+1)-PSI(I,J-1))/
1460      *      /2./DX
1470          NMID=X(I)+DX/2.
1480          VMID=Y(J)+DY/2.
1490          ATHETA=ATHETA-G*(NMID*DPSIDY+VMID**2+DX*DY-
1500      *      G*(NMID*DPSIDX+VMID**2+DX*DY)
1510      160 CONTINUE
1520      WRITE(6,100) ATHETA,IT
1530      170 FORMAT(1X,5N,1G,1ATHETA=,E11.5,5N,1TEPATIONS=,14.1)
1540      STOP
1550      END

```

APPENDIX B

SINGLE RING DEFLECTION PROGRAM

APPENDIX B

Single Ring Deflection Program

Program Description

The ring deflection program calculates the response of a single ring to given arbitrary loads and boundary conditions using the ring finite element discussed in Chapter 5. It models the ring with an even number of nodes and beam elements. Six degrees of freedom are allowed at each node. Concentrated and distributed loads can be applied in all six degrees. The program computes the deflection at each node in all six degrees of freedom, the Fourier coefficients of the deflections in the x- and y-directions and the twist about the θ -axis, the forces at the ends of each element, and the reaction forces at the supports.

Program Input

There must be an even number of nodes. The program is set for thirty nodes. They are numbered clockwise looking in the positive y-direction (Fig. B1).

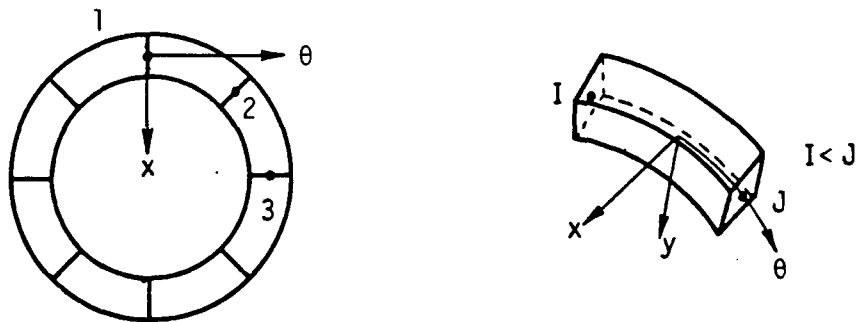


Figure B1. Node Numbering.

The program reads the input data from a data file which must have the following format:

I CONTROL	CARD (2I5)
columns	entry
1-5	Number of Nodes (setup is for 30) Must be even number of nodes evenly spaced.
6-10	Number of Fourier coefficients to be calculated (LE.8)
II ELEMENT	PROPERTIES (6F 10.0)
columns	entry
1-10	Radius at cross section centroid
11-20	a-cross sectional area
21-30	J_x --Moment of inertia for a circular beam about x-axis
31-40	J_y --Moment of inertia for a circular beam about y-axis
41-50	J_{xy} --Product of inertia for a circular beam
51-60	J_θ --Section torsional constant
III MATERIAL	PROPERTIES (2F10.0)
columns	entry
1-10	Young's modulus
11-20	Poisson's ratio

IV LOAD AND CONSTRAINT INFORMATION (2I5)

columns	entry
1-5	Number of nodes where concentrated loads are applied
6-10	Number of constrained nodes

V CONCENTRATED LOADS (I5, 6F10.0)

Input one card for each node where concentrated loads are applied.

columns	entry
1-5	Node number
6-15	Applied load in x-direction
16-25	Applied load in y-direction
26-35	Applied load in θ -direction
36-45	Applied moment about x-axis
46-45	Applied moment about y-axis
56-65	Applied moment about θ -axis

VI DISTRIBUTED LOADS (6F10.0)

This card must be input. If there are no distributed loads, enter all 0.0's.

columns	entry
1-10	Load in x-direction (total load)
11-20	Load in y-direction (total load)
21-30	Load in θ -direction (total load)
31-40	Moment about x-axis (total load)
41-50	Moment about y-axis (total load)
51-60	Moment about θ -axis (total load)

VII BOUNDARY CONDITIONS (CONSTRAINTS) (15, 6F10.0)

Input one card for each node where
boundary conditions are specified.

columns	entry
1-5	Node number
6-15	Constraint in x-direction Eq. 0.0 no constraint Eq. 1.0 no displacement allowed
16-25	Constraint in y-direction
26-35	Constraint in θ -direction
36-45	Constraint about x-axis
46-55	Constraint about y-axis
56-65	Constraint about θ -axis

Sample Problem

The program is used to model a ring with a one square inch cross
section and a 10-inch radius. The section properties are:

$$J_x = 0.0833 \text{ in}^4$$

$$J_y = 0.0833 \text{ in}^4$$

$$J_{xy} = 0.050 \text{ in}^4$$

$$J_\theta = 0.0982 \text{ in}^4$$

The ring is broken into a mesh of 30 nodes and elements. Concentrated loads of 50 lbs each were applied in the x-direction at nodes 8 and 23. A distributed load of 100 lbs total was applied in the y-direction (Fig. B2). Node 1 was constrained in the x-, y-, and θ -directions and node 16 was constrained in the y-direction. The data file, output, and program follow.

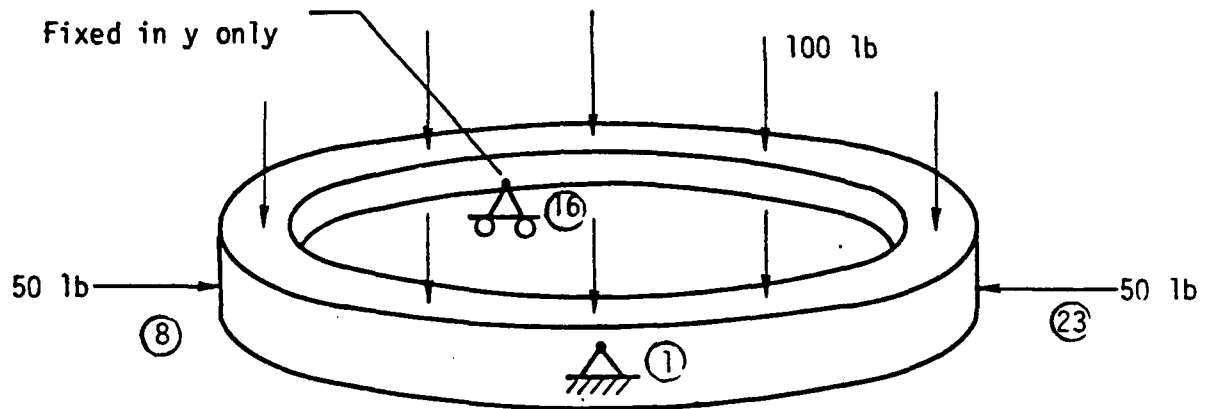


Figure B2. Applied Loads.

10	20	8						
20	10.0		1.0	0.0833	0.0833	0.05	0.0982	
30	0.0E06		0.1					
40	2	2						
50	0		50.0					
60	23		50.0					
70			100.0					
80	1		1.0	1.0	1.0	0.0	0.0	0.0
90	16		0.0	1.0	0.0	0.0	0.0	0.0
100								

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0
WARNING- AT NODE 1 BOTH LOAD AND BC ARE SPECIFIED
WARNING- AT NODE 15 BOTH LOAD AND BC ARE SPECIFIED

DISPLACEMENTS

N	U	V	W	DVDT	PSI	PHI
1	0.0	0.0	0.0	-0.96230-05	0.67150-04	0.41550-01
2	0.14240-03	0.19560-01	0.17610-04	0.18530-03	0.71930-04	0.36950-01
3	0.31520-02	0.72730-01	0.59650-04	0.20860-03	0.59610-04	0.26640-00
4	0.53420-02	0.14190-02	0.14670-03	0.15920-03	0.12860-03	0.10190-01
5	0.79800-03	0.21880-02	0.28310-03	0.24650-03	0.14500-03	-0.67350-04
6	0.99980-03	0.28570-02	0.46950-03	0.26630-03	0.13150-03	-0.22060-02
7	0.11110-02	0.33710-02	0.63210-03	0.20280-03	0.30140-04	-0.33540-00
8	0.18070-02	0.37030-02	0.91760-03	0.11690-03	-0.27400-04	-0.40290-01
9	0.61820-03	0.38580-02	0.10910-02	0.15350-04	-0.13070-03	-0.41850-00
10	0.54010-04	0.37490-02	0.11620-02	-0.11490-03	-0.17140-03	-0.37410-01
11	-0.54290-03	0.33680-02	0.11090-02	-0.24670-03	-0.16210-03	-0.37170-01
12	-0.10530-02	0.27350-02	0.91840-03	-0.35080-03	-0.11340-03	-0.33210-01
13	-0.14130-02	0.19050-02	0.67510-03	-0.40210-03	-0.56200-04	0.34190-04
14	-0.15720-02	0.11010-02	0.75810-03	-0.38120-03	0.63100-05	0.21930-01
15	-0.15450-02	0.29630-01	0.27700-04	-0.27710-03	0.51570-04	0.36840-00
16	-0.17640-02	0.0	-0.29170-03	-0.38140-04	0.67250-04	0.40550-01
17	-0.11310-02	0.33160-04	-0.54970-03	0.10900-03	0.73980-04	0.40480-01
18	-0.32440-03	0.40790-03	-0.76890-03	0.22690-03	0.99620-04	0.29910-01
19	-0.41980-03	0.37710-03	-0.89460-03	0.23570-03	0.12960-03	0.15030-00
20	0.63420-04	0.16040-02	-0.30380-03	0.29190-03	0.14500-03	-0.90870-05
21	0.55180-03	0.21770-02	-0.56980-03	0.24750-03	0.13150-03	-0.15260-02
22	0.95130-03	0.26250-02	-0.71100-03	0.17770-03	0.30140-04	-0.26170-01
23	0.11110-02	0.33710-02	0.63210-03	0.20280-03	0.30140-04	-0.33540-00
24	0.10420-02	0.38580-02	-0.28370-03	0.14160-04	-0.11070-03	-0.34040-01
25	0.74960-03	0.30920-02	-0.56860-04	-0.30630-04	-0.17140-03	-0.29950-01
26	0.29190-03	0.26880-02	0.51760-04	-0.20740-03	-0.16210-03	-0.20170-02
27	0.77500-04	0.21510-02	0.98500-04	-0.29910-03	-0.11940-03	-0.61740-04
28	-0.12760-03	0.14740-02	0.89810-04	-0.33950-03	-0.56100-04	0.10010-01
29	-0.19210-03	0.78190-03	0.52930-04	-0.10950-03	0.68110-05	0.19090-01
30	-0.11260-03	0.23160-03	0.15570-04	-0.20010-03	0.51570-04	0.36420-01

FOURIER COEFFICIENTS

N	U	V	PHI
1	0.7061E-01	0.1926E-01	0.1926E-04
2	0.3541E-01	0.1718E-02	0.4044E-01
3	0.1634E-02	0.1949E-03	0.1309E-03
4	0.1138E-01	0.1465E-01	0.2619E-04
5	0.5887E-03	0.1797E-03	0.1671E-03
6	0.1013E-04	0.1372E-04	0.1195E-05
7	0.1577E-03	0.1741E-03	0.9777E-03
8	0.7522E-05	0.7631E-05	0.1435E-05

FORCES AND MOMENTS

N	VX	VY	NT	MX	MY	MT
1	-0.26110-01	-0.21110-02	0.24850-02	-0.15860-01	-0.39580-02	0.12010-05
2	0.77250-01	0.11110-02	-0.11780-02	0.13660-03	0.73690-02	-0.17370-02
3	-0.77250-01	-0.10000-02	0.11780-02	-0.13660-03	-0.73690-02	0.27370-02
4	0.11780-01	0.10000-02	-0.11780-02	0.13660-03	0.73690-02	-0.40050-02

3	-0.12500+02	-0.16670+02	0.21650+02	-0.56190+02	-0.57440+02	0.45050+02
4	0.16710+02	0.16670+02	-0.13530+02	0.11610+02	0.26720+02	-0.52250+02
4	-0.16710+02	-0.16670+02	0.13530+02	-0.11610+02	-0.26720+02	0.52250+02
5	0.20210+02	0.16670+02	-0.14690+02	-0.27210+02	-0.12120+02	-0.50620+02
5	-0.20210+02	-0.16670+02	0.14690+02	0.27210+02	0.12120+02	0.50620+02
6	0.22840+02	0.16670+02	-0.10170+02	-0.57910+02	-0.57190+02	-0.41670+02
6	-0.22840+02	-0.16670+02	0.10170+02	0.57910+02	0.57190+02	0.41670+02
7	0.24450+02	0.16670+02	-0.51980+01	-0.79190+02	-0.10710+02	-0.27250+02
7	-0.24450+02	-0.16670+02	0.51980+01	0.79190+02	0.10710+02	0.27250+02
8	0.25000+02	0.16670+02	-0.10050+01	-0.90050+02	-0.10710+02	-0.27250+02
8	-0.25000+02	-0.16670+02	0.10050+01	0.90050+02	0.10710+02	0.27250+02
9	0.24450+02	0.16670+02	-0.51980+01	-0.79190+02	-0.10710+02	-0.27250+02
9	-0.24450+02	-0.16670+02	0.51980+01	0.79190+02	0.10710+02	0.27250+02
10	0.22840+02	0.16670+02	-0.10170+02	-0.57910+02	-0.57190+02	-0.41670+02
10	-0.22840+02	-0.16670+02	0.10170+02	0.57910+02	0.57190+02	0.41670+02
11	0.20210+02	0.16670+02	-0.14690+02	-0.27210+02	-0.12120+02	-0.50620+02
11	-0.20210+02	-0.16670+02	0.14690+02	0.27210+02	0.12120+02	0.50620+02
12	0.16710+02	0.16670+02	-0.13530+02	0.11610+02	0.26720+02	-0.52250+02
12	-0.16710+02	-0.16670+02	0.13530+02	-0.11610+02	-0.26720+02	0.52250+02
13	0.12500+02	0.16670+02	0.21650+02	-0.56190+02	-0.57440+02	0.45050+02
13	-0.12500+02	-0.16670+02	-0.21650+02	0.56190+02	0.57440+02	-0.45050+02
14	0.77250+01	0.20000+02	0.21730+02	-0.56890+02	-0.70690+02	0.27370+02
14	-0.77250+01	-0.20000+02	-0.21730+02	0.56890+02	0.70690+02	-0.27370+02
15	0.26110+01	0.21110+02	-0.24860+02	0.10660+02	0.39560+02	-0.12120+02
15	-0.26110+01	-0.21110+02	0.24860+02	-0.10660+02	-0.39560+02	0.12120+02
16	0.77250+01	0.20000+02	0.21730+02	-0.56890+02	-0.70690+02	0.27370+02
16	-0.77250+01	-0.20000+02	-0.21730+02	0.56890+02	0.70690+02	-0.27370+02
17	0.12500+02	0.16670+02	0.21650+02	-0.56190+02	-0.57440+02	0.45050+02
17	-0.12500+02	-0.16670+02	-0.21650+02	0.56190+02	0.57440+02	-0.45050+02
18	0.16710+02	0.16670+02	-0.13530+02	0.11610+02	0.26720+02	-0.52250+02
18	-0.16710+02	-0.16670+02	0.13530+02	-0.11610+02	-0.26720+02	0.52250+02

```

19 -0.16730+02 -0.11230+02 0.18580+02 -0.11510+02 -0.26720+02 0.52250+01
20 0.20230+02 0.10000+02 -0.14690+02 -0.27210+02 -0.12120+02 -0.50620+02

20 -0.20230+02 -0.10000+02 0.14690+02 0.27210+02 0.12120+02 0.50620+02
21 0.22840+02 0.10000+02 -0.10170+02 -0.57930+02 -0.57190+02 -0.41670+02

21 -0.22840+02 -0.66670+01 0.10170+02 0.57930+02 0.57190+02 0.41670+02
22 0.24450+02 0.66670+01 -0.51980+01 -0.79190+02 -0.10710+02 -0.27250+02

22 -0.24450+02 -0.66670+01 0.51980+01 0.79190+02 0.10710+02 0.27250+02
23 0.25000+02 0.33330+01 -0.10110+04 -0.90060+02 -0.15910+02 0.94650+01

23 0.25000+02 -0.44840+11 0.10110+04 0.90060+02 0.15910+02 0.94650+01
24 -0.24450+02 0.44840+11 -0.51980+01 -0.90060+02 -0.10710+02 0.94650+01

24 0.24450+02 0.33330+01 0.51980+01 0.90060+02 0.10710+02 -0.94650+01
25 -0.22840+02 -0.33330+01 -0.10170+02 -0.79190+02 -0.57190+02 0.41670+02

25 0.22840+02 0.66670+01 0.10170+02 0.79190+02 0.57190+02 -0.41670+02
26 -0.20230+02 -0.66670+01 -0.14690+02 -0.57930+02 -0.12120+02 -0.50620+02

26 0.20230+02 0.10000+02 0.14690+02 0.57930+02 0.12120+02 -0.50620+02
27 -0.16730+02 -0.10000+02 -0.18580+02 -0.17210+02 0.26720+02 0.52250+02

27 0.16730+02 0.11230+02 0.18580+02 0.27210+02 -0.16720+02 -0.52250+02
28 -0.11530+02 -0.11230+02 -0.11530+02 0.11510+02 -0.57440+02 -0.51250+02

28 0.11530+02 0.16670+02 0.11650+02 -0.11510+02 -0.57440+02 -0.51250+02
29 -0.77250+01 -0.16670+02 -0.21730+02 0.56390+02 0.78690+02 0.45050+02

29 0.77250+01 0.20000+02 0.21730+02 -0.56390+02 -0.78690+02 -0.45050+02
30 -0.26110+01 -0.20000+02 -0.24860+02 0.10660+02 0.39560+02 0.27870+02

30 0.26110+01 0.21230+02 0.24860+02 -0.10660+02 -0.39560+02 -0.27870+02
1 0.26110+01 -0.21230+02 -0.24860+02 0.10660+02 0.39560+02 -0.11230+05

```

RECTIONS AT CONSTRAINED NODES

```

N      R1      R2      R3      R4      R5      R6
1  0.1914E-05 -0.3000E-01 -0.1677E-05 0.1940E-05 0.7512E-05 -0.7491E-11
12 -0.2767E-05 -0.3000E-02 0.1101E-04 -0.5274E-05 -0.5610E-05 -0.2553E-11

```

STOP
TIME 0.4 SECS


```

10 C SINGLE RING DEFLECTION PROGRAM
20 C THIS PROGRAM DETERMINES THE REACTIONS OF A SINGLE RING
30 C SUBJECTED TO GIVEN LOADS AND BOUNDARY CONDITIONS. THE
40 C LOADS CAN BE NONAXISYMMETRIC. THERE MUST BE AN EVEN
50 C NUMBER OF NODES AND THEY MUST BE SPACED EVENLY. THEY
60 C ARE NUMBERED CLOCKWISE LOOKING IN THE POSITIVE Y-DIREC-
70 C TION. THIS PROGRAM COMPUTES THE DEFLECTIONS, THE
80 C FOURIER COEFFICIENTS OF THE DEFLECTIONS, THE FORCES IN
90 C THE RING, AND THE REACTION FORCES AT THE SUPPORTS.
100 C
110 DOUBLE PRECISION KP,AMAT,IMAT,IIMAT,IIIMAT,IVMAT,ENAT
120 DOUBLE PRECISION PIE,SUM(6,6),DELTA(12),FORCE(12)
130 REAL*4 JX,JY,JXY,JT,NT,MX,MY,MT,MXDL,MYDL,MTDL
140 DIMENSION CMX(20),CMY(20),CNT(20),CMX(20),CMY(20),CNT(20)
150 DIMENSION AMAT(180,25),IMAT(6,6),IIMAT(6,6),IIIMAT(6,6)
160 DIMENSION KP(12,12),IVMAT(6,6),ENAT(180)
170 DIMENSION VX(20),VY(20),NT(20),MX(20),MY(20),MT(20)
180 DIMENSION IIC(20),ICOMP(20),ICC(20),ICCOLD(20)
190 DIMENSION DDN(20),DDY(20),DDT(20),DDD(20),DDDY(20)
200 DIMENSION DDDT(20),DDD(20,6)
210 DIMENSION UGC(8,20),UGE(8,20),UC(8),VGC(8,20),VGE(8,20)
220 DIMENSION V(8),PGC(8,20),PGE(8,20),P(8)
230 DIMENSION REAC(20,5),OFOR(6),OFOR1(6),IFLAG(20),JFLAG(20)
240 CALL OPSYS ('ALLOCC','RDATA',3)
250 C
260 C READ PARAMETERS AND INITIALIZE VARIABLES
270 READ (9,2000) N,NCEF
280 READ (9,2010) RC,R,JX,JY,JXY,JT
290 READ (9,2020) E,PR
300 DO 100 I=1,N
310 VX(I)=0.0
320 VY(I)=0.0
330 NT(I)=0.0
340 MX(I)=0.0
350 MY(I)=0.0
360 MT(I)=0.0
370 DDN(I)=0.0
380 DDY(I)=0.0
390 DDT(I)=0.0
400 DDD(I)=0.0
410 DDDY(I)=0.0
420 DDDT(I)=0.0
430 IFLAG(I)=0
440 JFLAG(I)=0
450 100 CONTINUE
460 READ (9,2000) NCL,NEC
470 IF (NCL.EQ.0) GO TO 109
480 DO 105 I=1,NCL
490 READ (9,2040) ICL,VC(IND),VY(IND),NT(IND),MX(IND),MY(IND),MT(IND)
500 105 CONTINUE
510 109 READ (9,2010) MDL,YDL,TDL,MXDL,MYDL,MTDL
520 DO 106 I=1,N
530 VX(I)=VX(I)+NCL*V
540 VY(I)=VY(I)+YDL*V
550 NT(I)=NT(I)+TDL*V
560 MX(I)=MX(I)+MXDL*V
570 MY(I)=MY(I)+MYDL*V
580 MT(I)=MT(I)+MTDL*V
590 106 CONTINUE
600 DO 107 I=1,NEC
610 READ (9,2040) IND1,IC(IND1),DDN(IND1),DDY(IND1),DDT(IND1),DDD(IND1),DDDY(IND1)

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620      *          DDOT(IK)
620      IFLAG(I)=IN
640      JFLAG(IK)=I
650      107 CONTINUE
660      108 CONTINUE
670      2000 FORMAT (2I5)
680      2010 FORMAT (6F10.0)
690      2020 FORMAT (2F10.0)
700      2040 FORMAT (I5,6F10.0)
710 C
720      PIE=4.*DATAN(1.D+00)
730      PIES=PIE
740      DT=2.*PIE/N
750      G=E/2.0/(1.0+PR)
760 C
770 C CALCULATE INDEX RELATIONSHIPS - RENUMBERS NODES FOR COMPUTATIONAL
780 C EFFICIENCY
790      III(1)=1
800      III(2)=2
810      DO 110 I=3,N
820          AI=I
830          BI=I/2
840          AI=AI/2.-BI
850          IF(AI.GT.0.1) GO TO 120
860          III(I)=(I+2)/2
870          GO TO 110
880      110      III(I)=(2*N+3-I)/2
890      110 CONTINUE
900      DO 120 I=1,N
910          IIN=III(I)
920      120      ICOMP(IIN)=I
930 C
940 C RESECTION COMPUTATIONAL SUBSCRIPTS TO LOADS AND BOUNDARY CONDITIONS
950      II=1
960      II=III(I)
970      CMX(I)=XX(II)
980      CMY(I)=YY(II)
990      CNT(I)=NT(II)
1000      CMX(I)=MX(II)
1010      CMY(I)=MY(II)
1020      CNT(I)=MT(II)
1030      CDD(1,1)=DDX(II)
1040      CDD(1,2)=DDY(II)
1050      CDD(1,3)=DDT(II)
1060      CDD(1,4)=DDRX(II)
1070      CDD(1,5)=DDRY(II)
1080      CDD(1,6)=DDRT(II)
1090      150 CONTINUE
1100 C
1110 C CONSTRUCT THE STIFFNESS MATRIX
1120      CALL ALOCIPED A,3,6,6,6,6,6,6,6,6
1130      NEI=9*N
1140      NEAND=17
1150      DO 250 I=1,NEI
1160          EMAT(I)=0
1170          DO 250 J=1,NEI
1180      150      AMAT(I,J)=0
1190          DO 125 I=1,6
1200              DO 125 J=1,6
1210                  IMAT(I,J)=0
1220                  IIMAT(I,J)=0
1230                  IIMAT(I,J)=0
1240                  IVMAT(I,J)=0
1250      125 CONTINUE
1260      CALL ISEBKP IMAT
1270      CALL IVSUBKP IVMAT

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1190      CALL ADD(I,IMAT,I,MMAT,ISUM)
1195      CALL IISUB(KF,I,IMAT)
1200      CALL IISUB(KF,I,IMAT)
1210      J1=19
1220      DO 300 I=1,6
1230          J1=J1-1
1240          J2=J1+5
1250          JJ=0
1260          DO 300 J=J1,J2
1270              JJ=JJ+1
1280              AMAT(I,J)=SUM(I,JJ)
1290              AMAT(I,J-5)=IIMAT(I,JJ)
1300              IF((J+12).GT.35) GO TO 1301
1310              AMAT(I,J+12)=IIMAT(I,JJ)
1320              GO TO 300
1330      1301      WRITE(6,302)
1340      300      CONTINUE
1350      302      FORMAT (1X,'LINE 302 - CHECK COUNTERS J1 & J2')
1360      J1=19
1370      DO 400 I=7,12
1380          II=I-6
1390          J1=J1-1
1400          J2=J1+5
1410          JJ=0
1420          DO 400 J=J1,J2
1430              JJ=JJ+1
1440              AMAT(I,J)=SUM(II,JJ)
1450              AMAT(I,J+12)=IIMAT(II,JJ)
1460      400      AMAT(I,J-5)=IIMAT(II,JJ)
1470      NM2=N-2
1480      NODD=1
1490      DO 1000 NN=D,NM2
1500          IST=(NN-1)*6+1
1510          ISTOP=IST+5
1520          IF NODD.EQ.1, GO TO 1550
1530          GO TO 1560
1540      1050      CONTINUE
1550      1060      J1=19
1560          II=0
1570          DO 1070 I=IST,ISTOP
1580              II=II+1
1590              J1=J1-1
1600              J2=J1+5
1610              JJ=0
1620              DO 1070 J=J1,J2
1630                  JJ=JJ+1
1640                  AMAT(I,J)=SUM(II,JJ)
1650                  GO TO (1080,1090),NODD
1660      1080          IF (J-12).LT.1, GO TO 1081
1670                  AMAT(I,J-12)=IIMAT(II,JJ)
1680      1091          IF (J-12).GT.35) GO TO 1071
1690                  AMAT(I,J-12)=IIMAT(II,JJ)
1700                  GO TO 1070
1710      1070          IF (J-12).LT.1, GO TO 1091
1720                  AMAT(I,J-12)=IIMAT(II,JJ)
1730      1091          IF (J+12).GT.35) GO TO 1071
1740                  AMAT(I,J+12)=IIMAT(II,JJ)
1750      1070      CONTINUE
1760      GO TO 1072
1770      1071      WRITE(6,1070)
1780      1072      FORMAT (1X,'LINE 1072 - CHECK COUNTERS J1 & J2')
1790      CONTINUE
1800      IF NODD.EQ.1, GO TO 1095
1810      NODD=1
1820      GO TO 1080
1830      1095      NODD=1

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1940 1000 CONTINUE
1950      NM1=N-1
1960      IST=NM1*6+1
1970      ISTOP=IST+5
1980      J1=19
1990      II=0
2000      DO 500 I=IST,ISTOP
2010          II=II+1
2020          J1=J1-1
2030          J2=J1+5
2040          JJ=0
2050          DO 500 J=J1,J2
2060              JJ=JJ+1
2070              AMAT(I,J)=SUM(II,JJ)
2080              IF((J-12).LT.1) GO TO 500
2090              AMAT(I,J-12)=IIMAT(II,JJ)
2100          500      AMAT(I,J+6)=IIMAT(II,JJ)
2110      IST=NM1*6+1
2120      ISTOP=IST+5
2130      J1=19
2140      II=0
2150      DO 600 I=IST,ISTOP
2160          II=II+1
2170          J1=J1-1
2180          J2=J1+5
2190          JJ=0
2200          DO 600 J=J1,J2
2210              JJ=JJ+1
2220              AMAT(I,J)=SUM(II,JJ)
2230              IF((J-12).LT.1) GO TO 600
2240              AMAT(I,J-12)=IIMAT(II,JJ)
2250          600      AMAT(I,J+6)=IIMAT(II,JJ)
2260 C
2270 C CONSTRUCT THE FORCE MATRIX
2280      DO 700 NR=1,N
2290          NC=(NR-1)*6+1
2300          BMAT(NC)=BMAT(NC)+CVX(NR)
2310          BMAT(NC+1)=BMAT(NC+1)+CVY(NR)
2320          BMAT(NC+2)=BMAT(NC+2)+CNT(NR)
2330          BMAT(NC+3)=BMAT(NC+3)+CMX(NR)
2340          BMAT(NC+4)=BMAT(NC+4)+CMY(NR)
2350          BMAT(NC+5)=BMAT(NC+5)+CMT(NR)
2360      700 CONTINUE
2370 C
2380 C INCLUDE BOUNDARY CONDITIONS IN STIFFNESS MATRIX AND
2390 C CHECK FOR NODES WITH BOTH LOAD AND BOUNDARY CONDITION SPECIFIED
2400      DO 750 I=1,N
2410          NA=III(I)
2420          DO 751 II=1,6
2430              IF(DDD(I,II).EQ.0.) GO TO 751
2440              JJJ=6*(I-1)+II
2450              DO 752 J=1,15
2460                  752      AMAT(JJJ,J)=0.
2470                  AMAT(JJJ,18)=1.
2480                  IF(EMAT(JJJ).EQ.0.) GO TO 751
2490                  NR=JFLAG(NR)
2500                  REAC(NR,II)=REAC(NR,II)-BMAT(JJJ)
2510                  BMAT(JJJ)=0.0
2520                  WRITE(6,754) NA
2530          754      FORMAT (1X,'WARNING- AT NODE ',I4, ' BOTH LOAD AND BC ARE
2540              *          ' SPECIFIED')
2550          751      CONTINUE
2560      750 CONTINUE
2570 C
2580 C SOLVE FOR DISPLACEMENTS
2590      CALL GAUSS (AMAT,EMAT,REAC,NR)

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2600 C
2610 C PRINT DISPLACEMENTS
2620 WRITE(6,7)
2630 7 FORMAT(//1X,'DISPLACEMENTS',/3X,'NN',6X,'U',11X,'V',11X,'W',10X,
2640 * 'DVT',8X,'PSI',9X,'PHI')
2650 DO 800 NN=1,N
2660 IC=ICOMP(NN)
2670 IC=(IC-1)*6
2680 WRITE(6,5) NN,(BMAT(IC+J),J=1,6)
2690 5 FORMAT(1X,13,6E12,4)
2700 800 CONTINUE
2710 C
2720 C COMPUTE AND WRITE FOURIER COEFFICIENTS OF DEFLECTIONS
2730 THETA=0.0
2740 DO 802 NN=1,N
2750 IC=ICOMP(NN)
2760 IC=(IC-1)*6
2770 DO 801 INT=1,NCDEF
2780 UGC(INT,NN)=BMAT(IC-1)*COS(INT*THETA)
2790 UGS(INT,NN)=BMAT(IC+1)*SIN(INT*THETA)
2800 VGC(INT,NN)=BMAT(IC+2)*COS(INT*THETA)
2810 VGS(INT,NN)=BMAT(IC+2)*SIN(INT*THETA)
2820 PGC(INT,NN)=BMAT(IC+5)*COS(INT*THETA)
2830 PGS(INT,NN)=BMAT(IC+5)*SIN(INT*THETA)
2840 801 CONTINUE
2850 THETA=THETA+DT
2860 802 CONTINUE
2870 CALL FOUR(UGC,UGS,U,N,NCDEF)
2880 CALL FOUR(VGC,VGS,V,N,NCDEF)
2890 CALL FOUR(PGC,PGS,P,N,NCDEF)
2900 WRITE(6,202)
2910 202 FORMAT(//1X,'FOURIER COEFFICIENTS',/3X,'NN',8X,'U',11X,'V',
2920 * '10X,'PHI')
2930 203 1X,'NCDEF'
2940 WRITE(6,203) 1,2,3,4,5,6,7,8,9,10,11,12,13,14
2950 103 FORMAT(1X,13,6E12,4)
2960 204 CONTINUE
2970 C
2980 C WRITE FORCES AND MOMENTS
2990 WRITE(6,11)
3000 11 FORMAT(//1X,'FORCES AND MOMENTS')
3010 WRITE(6,12)
3020 12 FORMAT(//3X,'N',6X,'VX',10X,'VY',10X,'NT',10X,'MX',10X,'MY',10X,
3030 * 'MT')
3040 DO 5000 I1=1,N
3050 I2=I1+1
3060 IF(I2.GT.N) I2=1
3070 IC1=ICOMP(I1)-1)*6
3080 IC2=ICOMP(I2)-1)*6
3090 DO 5001 I=1,6
3100 DELTA(I)=BMAT(I1-I)
3110 DELTA(I+6)=BMAT(I2-I)
3120 5001 CONTINUE
3130 DO 5002 I=1,12
3140 FORCE(I)=0
3150 DO 5002 J=1,12
3160 5002 FORCE(I)=FORCE(I)+PP(I,J)*DELTA(J)
3170 WRITE(6,13) I1,(FORCE(J),J=1,6)
3180 WRITE(6,13) I2,(FORCE(J),J=7,12)
3190 13 FORMAT(1X,13,6E12,4)
3200 WRITE(6,14)
3210 14 FORMAT(//)
3220 IF(I1.NE.1) GO TO 5006
3230 IF(JFLAG(1).EQ.0) GO TO 5020
3240 DO 5005 IT=1,6
3250 OFOR(I1,IT)=FORCE(IT)

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3260 6005 CONTINUE
3270 GO TO 6020
3280 6006 IF(JFLAG(11).EQ.0) GO TO 6010
3290 NR=JFLAG(11)
3300 DO 6010 IT=1,6
3310 REAC(NR,IT)=REAC(NR,IT)+OFOR(IT)+FORCE(IT)
3320 6010 CONTINUE
3330 6020 IF(JFLAG(12).EQ.0) GO TO 6040
3340 DO 6020 IT=1,6
3350 OFOR(IT)=FORCE(IT+6)
3360 6020 CONTINUE
3370 6040 IF(12.EQ.1.AND.JFLAG(1).NE.0) GO TO 6050
3380 GO TO 6000
3390 6050 NR=JFLAG(1)
3400 DO 6060 IT=1,6
3410 REAC(NR,IT)=REAC(NR,IT)+OFOR1(IT)+FORCE(IT+6)
3420 6060 CONTINUE
3430 6000 CONTINUE
3440 WRITE (6,6090)
3450 DO 6070 I=1,NBC
3460 WRITE (6,6090) IFLAG(I), (REAC(I,J),J=1,6)
3470 6070 CONTINUE
3480 6080 FORMAT(//1X,'REACTIONS AT CONSTRAINED NODES',//1X,'N',
3490 * 5X,'RX',10X,'RY',10X,'RT',5X,'RMX',5X,'RMY',
3500 * 5X,'RMT')
3510 6090 FORMAT(1X,10,6E12,4)
3520 STOP
3530 END
3540 C
3550 C
3560 SUBROUTINE ISUB(KP,IMAT)
3570 DOUBLE PRECISION KP(12,12),IMAT(6,6)
3580 DO 10 I=1,6
3590 IMAT(I,J)=KP(I,J)
3600 10 RETURN
3610 END
3620 C
3630 C
3640 C
3650 SUBROUTINE IVSUB(KP,IVMAT)
3660 DOUBLE PRECISION KP(12,12),IVMAT(6,6)
3670 DO 10 I=1,6
3680 DO 10 J=1,6
3690 II=I+6
3700 JJ=J+6
3710 10 IVMAT(I,J)=KP(II,JJ)
3720 RETURN
3730 END
3740 C
3750 C
3760 SUBROUTINE IISUB(KP,IIMAT)
3770 DOUBLE PRECISION KP(12,12),IIMAT(6,6)
3780 DO 10 I=1,6
3790 DO 10 J=1,6
3800 JJ=J+6
3810 10 IIMAT(I,J)=KP(I,JJ)
3820 RETURN
3830 END
3840 C
3850 C
3860 SUBROUTINE IIISUB(KP,IIIMAT)
3870 DOUBLE PRECISION KP(12,12),IIIMAT(6,6)
3880 DO 10 I=1,6
3890 DO 10 J=1,6
3900 II=I+6
3910 10 IIIMAT(I,J)=KP(II,JJ)

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3920      RETURN
3930      END
3940 C
3950 C
3960      SUBROUTINE ADD(MATA, MATB, MATSUM)
3970      DOUBLE PRECISION MATA(6,6), MATB(6,6), MATSUM(6,6)
3980      DO 10 I=1,6
3990          DO 10 J=1,6
4000      10      MATSUM(I,J)=MATA(I,J)+MATB(I,J)
4010      RETURN
4020      END
4030 C
4040 C
4050      SUBROUTINE FOUR(GRANDC, GRANDS, Y, N, NCOEF)
4060      DIMENSION GRANDC(NCOEF,N), GRANDS(NCOEF,N), Y(NCOEF)
4070      DIMENSION A(S), B(S)
4080      PI=4.0*ATAN(1.0)
4090      DT=2*PI/N
4100      DO 20 I=1, NCOEF
4110          A(I)=0.0
4120          B(I)=0.0
4130          DO 10 J=1, N
4140              A(I)=A(I)+GRANDC(I,J)*DT/PI
4150              B(I)=B(I)+GRANDS(I,J)*DT/PI
4160      10      CONTINUE
4170          Y(I)=(A(I)**2+B(I)**2)**0.5
4180      20 CONTINUE
4190      RETURN
4200      END
4210 C
4220 C
4230      SUBROUTINE HOLCOP(RC, AREA, JX, JY, JXY, JT, E, G, DT, K)
4240      DOUBLE PRECISION RC(12,12), K(12,12), WORK(12), DET(20), D(12,12)
4250      DOUBLE PRECISION T(2), PIE, R, RA, JYOR2, JXYOJX, JXYOJY, P, Q, SS
4260      DOUBLE PRECISION U, V, S, C, TH, F
4270      DIMENSION IPVT(12)
4280      PIE=4.0*ATAN(1.0+0.0)
4290      T(1)=1
4300      T(2)=T(1)+DT
4310      R=RC
4320      RA=E+JX/G/JT
4330      JYOR2=JY/AREA/R/R
4340      JXYOJX=JXY/JX
4350      JXYOJY=JXY/JY
4360      P=1./JXYOJX
4370      Q=0.5*(1.+1./RA)/JXYOJX
4380      SS=(1.-JYOR2)/JXYOJX-JYOR2+JX+OJY
4390      U=1./JXYOJY-JXYOJX
4400      V=(1.-1./RA)/JXYOJY-JX+OJX
4410      DO 10 I=1,12
4420          DO 10 J=1,12
4430              A(I,J)=0
4440              K(I,J)=0
4450      10      D(I,J)=0
4460      F=1.
4470      DO 20 II=1,2
4480          TH=T(II)
4490          C=DCOS(TH)
4500          S=DSIN(TH)
4510          I=1+(II-1)*6
4520          A(I,20)=-Q*TH+C
4530          A(I,40)=Q*TH+S
4540          A(I,50)=-P*TH+S
4550

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4560      A(I,6)=-P*TH*C
4570      P(I,7)=P
4580      A(I,8)=-Q*TH*C
4590      A(I,9)=Q*TH*S
4600      A(I,10)=P*S
4610      A(I,11)=P*C
4620      I=I+1
4630      A(I,1)=1.
4640      A(I,2)=TH
4650      A(I,3)=S
4660      A(I,4)=C
4670      A(I,5)=TH*S
4680      A(I,6)=TH*C
4690      I=I+1
4700      A(I,3)=-Q*(C+TH*S)
4710      A(I,4)=Q*(S-TH*C)
4720      A(I,5)=-P*(S-TH*C)
4730      A(I,6)=-P*(C+TH*S)
4740      A(I,7)=SS*TH
4750      A(I,8)=-Q*(C+TH*S)
4760      A(I,9)=Q*(S-TH*C)
4770      A(I,11)=P*S
4780      A(I,10)=-P*C
4790      A(I,12)=1.
4800      I=I+1
4810      A(I,2)=1./R
4820      A(I,3)=C/R
4830      A(I,4)=-S/R
4840      A(I,5)=(TH*C+S)/R
4850      A(I,6)=(-TH*S+C)/R
4860      I=I+1
4870      A(I,2)=-2.*Q*C/R
4880      A(I,4)=2.*Q*S/R
4890      A(I,5)=-2.*P*S/R
4900      A(I,6)=-2.*P*C/R
4910      A(I,7)=2.*P*TH/R
4920      A(I,3)=-2.*Q*C/R
4930      A(I,9)=2.*Q*S/R
4940      A(I,12)=1./R
4950      I=I+1
4960      A(I,5)=-TH*S/R
4970      A(I,6)=-TH*C/R
4980      A(I,7)=1./R
4990      A(I,8)=S/R
5000      A(I,9)=C/R
5010      I=1+(II-1)*6
5020      D(I,3)=V*C+F/P
5030      D(I,4)=-V*S+F/R
5040      D(I,5)=2.*U*S+F/R
5050      D(I,6)=2.*U*C+F/R
5060      D(I,8)=V*C+F/R
5070      D(I,9)=-V*S+F/R
5080      I=I+1
5090      D(I,2)=-1.*F/R/AA
5100      I=I+1
5110      D(I,3)=V*S+F/R
5120      D(I,4)=V*C+F/R
5130      D(I,5)=-2.*U*C+F/R
5140      D(I,6)=2.*U*S+F/R
5150      D(I,8)=V*S+F/R
5160      D(I,9)=V*C+F/R
5170      I=I+1
5180      D(I,2)=-S*F/AA
5190      D(I,4)=-C*F/AA
5200      D(I,8)=-S*F/AA
5210      D(I,9)=-C*F/AA

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5220      I=I+1
5230      D(I,3)=-V*S*F
5240      D(I,4)=-V*C*F
5250      D(I,5)=2.*U*C*F
5260      D(I,6)=-2.*U*S*F
5270      D(I,7)=-U*F
5280      D(I,8)=-V*S*F
5290      D(I,9)=-V*C*F
5300      I=I+1
5310      D(I,2)=-F/AA
5320      D(I,3)=-C*F/AA
5330      D(I,4)=S*F/AA
5340      D(I,8)=-C*F/AA
5350      D(I,9)=S*F/AA
5360      F=-1.
5370 20 CONTINUE
5380      CALL DGEFA(A,12,12,IPVT,INFO)
5390      WRITE(6,1) INFO
5400      1 FORMAT(1H ,I5)
5410      JOB=1
5420      CALL DGED1(A,12,12,IPVT,DET,WORK,JOB)
5430      DO 30 I=1,12
5440          DO 30 J=1,12
5450              DO 40 JJ=1,12
5460                  40      K(I,J)=K(I,J)+D(I,JJ)*A(JJ,J)
5470                      K(I,J)=K(I,J)*E+JX/R**2
5480 20 CONTINUE
5490 2  FORMAT(1H ,6E12,4)
5500      RETURN
5510      END
5520 C
5530 C
5540      SUBROUTINE DAEFYN(DA,DX,INCX,DY,INCY)
5550      DOUBLE PRECISION DA,DX,INCX,DY,INCY
5560      INTEGER I,INCX,INCY,INI,M,MP1,N
5570      IF(N.LE.0)RETURN
5580      IF(DA.EQ.0.000) RETURN
5590      IF(INCX.EQ.1.AND.INCY.EQ.1)GO TO 20
5600      IX = 1
5610      IY = 1
5620      IF(INCX.LT.0)IX = (-N+1)*INCX + 1
5630      IF(INCY.LT.0)IY = (-N+1)*INCY + 1
5640      DO 10 I = 1,N
5650          DY(IY) = DY(IY) + DA*DX(IX)
5660          IX = IX + INCX
5670          IY = IY + INCY
5680 10 CONTINUE
5690      RETURN
5700 20 M = MOD(N,4)
5710      IF(M.EQ.0)GO TO 40
5720      DO 10 I = 1,M
5730          DY(I) = DY(I) + DA*DX(I)
5740 20 CONTINUE
5750      IF(N.LT.4)RETURN
5760 40 MP1 = M + 1
5770      DO 50 I = MP1,N,4
5780          DY(I) = DY(I) + DA*DX(I)
5790          DY(I + 1) = DY(I + 1) + DA*DX(I + 1)
5800          DY(I + 2) = DY(I + 2) + DA*DX(I + 2)
5810          DY(I + 3) = DY(I + 3) + DA*DX(I + 3)
5820 50 CONTINUE
5830      RETURN
5840      END
5850 C
5860 C
5870      INTEGER FUNCTION IDMAXN(DX,INCX)

```

```

5880      DOUBLE PRECISION DX(1), DMAX
5890      INTEGER I, INCX, IX, N
5900      IDAMAX = 0
5910      IF( N .LT. 1 ) RETURN
5920      IDAMAX = 1
5930      IF( N .EQ. 1 ) RETURN
5940      IF( INCX .EQ. 1 ) GO TO 20
5950      IX = 1
5960      DMAX = DABS( DX(1) )
5970      IX = IX + INCX
5980      DO 10 I = 2, N
5990          IF( DABS( DX(IX) ) .LE. DMAX ) GO TO 5
6000          IDAMAX = I
6010          DMAX = DABS( DX(IX) )
6020          IX = IX + INCX
6030      5 CONTINUE
6040      RETURN
6050      20 DMAX = DABS( DX(1) )
6060      DO 10 I = 2, N
6070          IF( DABS( DX(I) ) .LE. DMAX ) GO TO 20
6080          IDAMAX = I
6090          DMAX = DABS( DX(I) )
6100      10 CONTINUE
6110      RETURN
6120      END
6130 C
6140 C
6150      SUBROUTINE DSCAL( N, DA, DX, INCX )
6160      DOUBLE PRECISION DA, DX(1)
6170      INTEGER I, INCX, M, MP1, N, NINCX
6180      IF( N .LE. 0 ) RETURN
6190      IF( INCX .EQ. 1 ) GO TO 20
6200      NINCX = N * INCX
6210      DO 10 I = 1, NINCX, INCX
6220          DX(I) = DA * DX(I)
6230      10 CONTINUE
6240      RETURN
6250      20 M = MOD( N, 5 )
6260      IF( M .EQ. 0 ) GO TO 40
6270      DO 20 I = 1, M
6280          DX(I) = DA * DX(I)
6290      20 CONTINUE
6300      IF( N .LT. 5 ) RETURN
6310      40 MP1 = M + 1
6320      DO 50 I = MP1, N, 5
6330          DX(I) = DA * DX(I)
6340          DX(I + 1) = DA * DX(I + 1)
6350          DX(I + 2) = DA * DX(I + 2)
6360          DX(I + 3) = DA * DX(I + 3)
6370          DX(I + 4) = DA * DX(I + 4)
6380      50 CONTINUE
6390      RETURN
6400      END
6410 C
6420 C
6430      SUBROUTINE DSWAP( N, DX, INCX, DY, INCY )
6440      DOUBLE PRECISION DX(1), DY(1), DTEMP
6450      INTEGER I, INCX, INCY, IX, IY, M, MP1, N
6460      IF( N .LE. 0 ) RETURN
6470      IF( INCX .EQ. 1 .AND. INCY .EQ. 1 ) GO TO 10
6480      IX = 1
6490      IY = 1
6500      IF( INCX .LT. 0 ) IX = (-N+1) * INCX + 1
6510      IF( INCY .LT. 0 ) IY = (-N+1) * INCY + 1
6520      DO 10 I = 1, N
6530          DTEMP = DX(IX)

```

```

6540      DX(IY) = DY(IY)
6550      DY(IY) = DTEMP
6560      IX = IX + INCX
6570      IY = IY + INCY
6580 10 CONTINUE
6590      RETURN
6600 20 M = MOD(N,3)
6610      IF( M .EQ. 0 ) GO TO 40
6620      DO 30 I = 1,M
6630          DTEMP = DX(I)
6640          DX(I) = DY(I)
6650          DY(I) = DTEMP
6660 30 CONTINUE
6670      IF( N .LT. 3 ) RETURN
6680 40 MP1 = M + 1
6690      DO 50 I = MP1,N,3
6700          DTEMP = DX(I)
6710          DX(I) = DY(I)
6720          DY(I) = DTEMP
6730          DTEMP = DX(I + 1)
6740          DX(I + 1) = DY(I + 1)
6750          DY(I + 1) = DTEMP
6760          DTEMP = DX(I + 2)
6770          DX(I + 2) = DY(I + 2)
6780          DY(I + 2) = DTEMP
6790 50 CONTINUE
6800      RETURN
6810      END
6820 C
6830 C
6840      SUBROUTINE DGEFA(A,LDA,N,IPVT,INFO)
6850      INTEGER LDA,N,IPVT(1),INFO
6860      DOUBLE PRECISION A(LDA,1)
6870      DOUBLE PRECISION T
6880      INTEGER IDAMAX,J,K,KP1,L,NM1
6890      INFO = 0
6900      NM1 = N - 1
6910      IF (NM1 .LT. 1) GO TO 70
6920      DO 60 K = 1, NM1
6930          KP1 = K + 1
6940          L = IDAMAX(N-K+1,A(K,K),1) + K - 1
6950          IPVT(K) = L
6960          IF (A(L,K) .EQ. 0.0D0) GO TO 40
6970          IF (L .EQ. K) GO TO 10
6980          T = A(L,K)
6990          A(L,K) = A(K,K)
7000          A(K,K) = T
7010 10 CONTINUE
7020          T = -1.0D0/A(K,K)
7030          CALL DSCAL(N-K,T,A(K+1,K),1)
7040          DO 20 J = KP1, N
7050              T = A(L,J)
7060              IF (L .EQ. K) GO TO 10
7070              A(L,J) = A(K,J)
7080              A(K,J) = T
7090 20 CONTINUE
7100          CALL DAXPY(N-K,T,A(K+1,K),1,A(K+1,J),1)
7110 30 CONTINUE
7120          GO TO 50
7130 40 CONTINUE
7140          INFO = K
7150 50 CONTINUE
7160 60 CONTINUE
7170 70 CONTINUE
7180          IPVT(N) = N
7190          IF (A(N,N) .EQ. 0.0D0) INFO = N

```

```

7200      RETURN
7210      END
7220 C
7230 C
7240      SUBROUTINE DGEDI(A, LDA, N, IPVT, DET, WORK, JOB)
7250      INTEGER LDA, N, IPVT(1), JOB
7260      DOUBLE PRECISION A(LDA, 1), DET(2), WORK(1)
7270      DOUBLE PRECISION T
7280      DOUBLE PRECISION TEN
7290      INTEGER I, J, K, KB, KP1, L, NM1
7300      IF (JOB/10 .EQ. 0) GO TO 70
7310      DET(1) = 1.000
7320      DET(2) = 0.000
7330      TEN = 10.000
7340      DO 50 I = 1, N
7350          IF (IPVT(I) .NE. I) DET(1) = -DET(1)
7360          DET(1) = A(I, I)*DET(1)
7370          IF (DET(1) .EQ. 0.000) GO TO 60
7380      10      IF (DABS(DET(1)) .GE. 1.000) GO TO 20
7390          DET(1) = TEN*DET(1)
7400          DET(2) = DET(2) - 1.000
7410          GO TO 10
7420      20      CONTINUE
7430      30      IF (DABS(DET(1)) .LT. TEN) GO TO 40
7440          DET(1) = DET(1)/TEN
7450          DET(2) = DET(2) + 1.000
7460          GO TO 30
7470      40      CONTINUE
7480      50      CONTINUE
7490      60      CONTINUE
7500      70      CONTINUE
7510      IF (MOD(JOB, 10) .EQ. 0) GO TO 150
7520      DO 100 K = 1, N
7530          A(K, 1) = 1.000/A(K, K)
7540          T = -A(K, K)
7550          CALL DSCAL(K-1, T, A(1, K), 1)
7560          KP1 = K + 1
7570          IF (N .LT. KP1) GO TO 90
7580          DO 80 J = KP1, N
7590              T = A(K, J)
7600              A(K, J) = 0.000
7610              CALL DAXPY(K, T, A(1, K), 1, A(1, J), 1)
7620      80      CONTINUE
7630      90      CONTINUE
7640      100     CONTINUE
7650      NM1 = N - 1
7660      IF (NM1 .LT. 1) GO TO 140
7670      DO 110 KB = 1, NM1
7680          L = N - KB
7690          KP1 = K + 1
7700          DO 110 I = KP1, N
7710              WORK(I) = A(I, K)
7720              A(I, K) = 0.000
7730      110     CONTINUE
7740          DO 120 J = KP1, N
7750              T = WORK(J)
7760              CALL DAXPY(N, T, A(1, J), 1, A(1, K), 1)
7770      120     CONTINUE
7780          L = IPVT(K)
7790          IF (L .NE. K) CALL DSWAP(N, A(1, K), 1, A(1, L), 1)
7800      130     CONTINUE
7810      140     CONTINUE
7820      150     CONTINUE
7830      RETURN
7840      END
7850 C

```

```

7860 C
7870 SUBROUTINE GAUSS(A,B,MBAND,NEQ)
7880 C THIS SUBPROGRAM PERFORMS GAUSSIAN ELIMINATION
7890 C ON A NON SYMMETRICAL Banded MATRIX
7900 DOUBLE PRECISION A(6300),B(180),C,S
7910 NQ1=NEQ-1
7920 DO 10 I=1,NQ1
7930 I1=I+1
7940 I2=I+MBAND
7950 DO 20 J1=I1,I2
7960 J=J1
7970 K=(J-I1+MBAND)*NEQ+I1
7980 IF(A(K).EQ.0.) GO TO 20
7990 KK=(J-I+MBAND)*NEQ+I
8000 C=A(K)/A(KK)
8010 J1=J
8020 J2=J+MBAND
8030 IF(J2.GT.NEQ) J2=NEQ
8040 DO 40 JJ=J1,J2
8050 KKK=(JJ-I1+MBAND)*NEQ+I1
8060 KKKK=(JJ-I+MBAND)*NEQ+I
8070 40 A(KKK)=A(KKK)-C*A(KKKK)
8080 B(I1)=B(I1)-C*B(I)
8090 20 CONTINUE
8100 10 CONTINUE
8110 K=MBAND*NEQ+NEQ
8120 B(NEQ)=B(NEQ)/A(K)
8130 DO 50 I1=1,NQ1
8140 I=NEQ-I1
8150 J1=I+1
8160 J2=I+MBAND
8170 S=0.
8180 IF(J2.GT.NEQ) J2=NEQ
8190 DO 60 J=J1,J2
8200 60 S=S-A(K)*B(J)
8210 KK=MBAND*NEQ+I
8220 50 B(I)=(B(I)-S)/A(KK)
8230 RETURN
8240 END
8250

```

APPENDIX C
TWO RING DEFLECTION PROGRAM

APPENDIX C

Two Ring Deflection Program

Program Description

The two ring deflection program calculates the face deflections and forces developed in a two ring face seal subjected to operating loads. It uses the finite element method to model the seal with beam and spring elements.

The program models the rings with an even number of nodes and beam elements. As presented here, it assumes even spacing of the nodes so each element is $2\pi R/(\# \text{ of nodes})$ long. Variable element size can also be used.

Spring elements connect the nodes on the primary ring to corresponding nodes on the mating ring. These model the face contact. Where the ring faces touch, the springs' stiffness is set equal to the stiffness of the carbon insert. The springs have zero stiffness where there is no contact. As the load is applied and the rings deform, springs are given the face stiffness or zero stiffness as they come into or go out of contact.

Both rings may have an initial face profile. The datum from which the profiles are measured is arbitrary since the program subtracts out the first harmonic tilt between the two rings. The program calculates a likely initial-face contact pattern after tilting the seal faces.

To avoid instability in the solution, the program applies the loads incrementally. After the total loads are applied, the program

iterates up to two more times to see that the contact pattern is constant. It prints a message confirming convergence if the solution is consistent. When the total loads have been applied, the deflections and forces are calculated and printed. The average gap and the cube mean gap are computed at the end.

Program Input

The two ring deflection program requires information about the ring geometries, ring materials, and applied forces. Some data is read from a file and some is input by changing program lines.

The program lines that contain input information are:

Line 310: N--Number of nodes (set up for 30)

Line 320: RCP--Radius to the centroid of the primary ring

Line 360: AP--Cross section area of the primary ring

Line 370: JXP--Moment of inertia about the x-axis of primary
ring

$$\left(JXP = \int_A \frac{y^2}{1 - x/R} dA \right)$$

Line 380: JYP--Moment of inertia about the y-axis of primary ring

Line 390: JXYP--Product of inertia of primary ring

Line 400: JTP--Torsional constant of primary ring cross section

Line 410: EP--Young's modulus of primary ring material

Line 420: GP--Shear modulus of primary ring material

Lines 430-500: RCM, AM, JXM, JYM, JXYM, JTM, EM, GM--Same as above for
mating ring

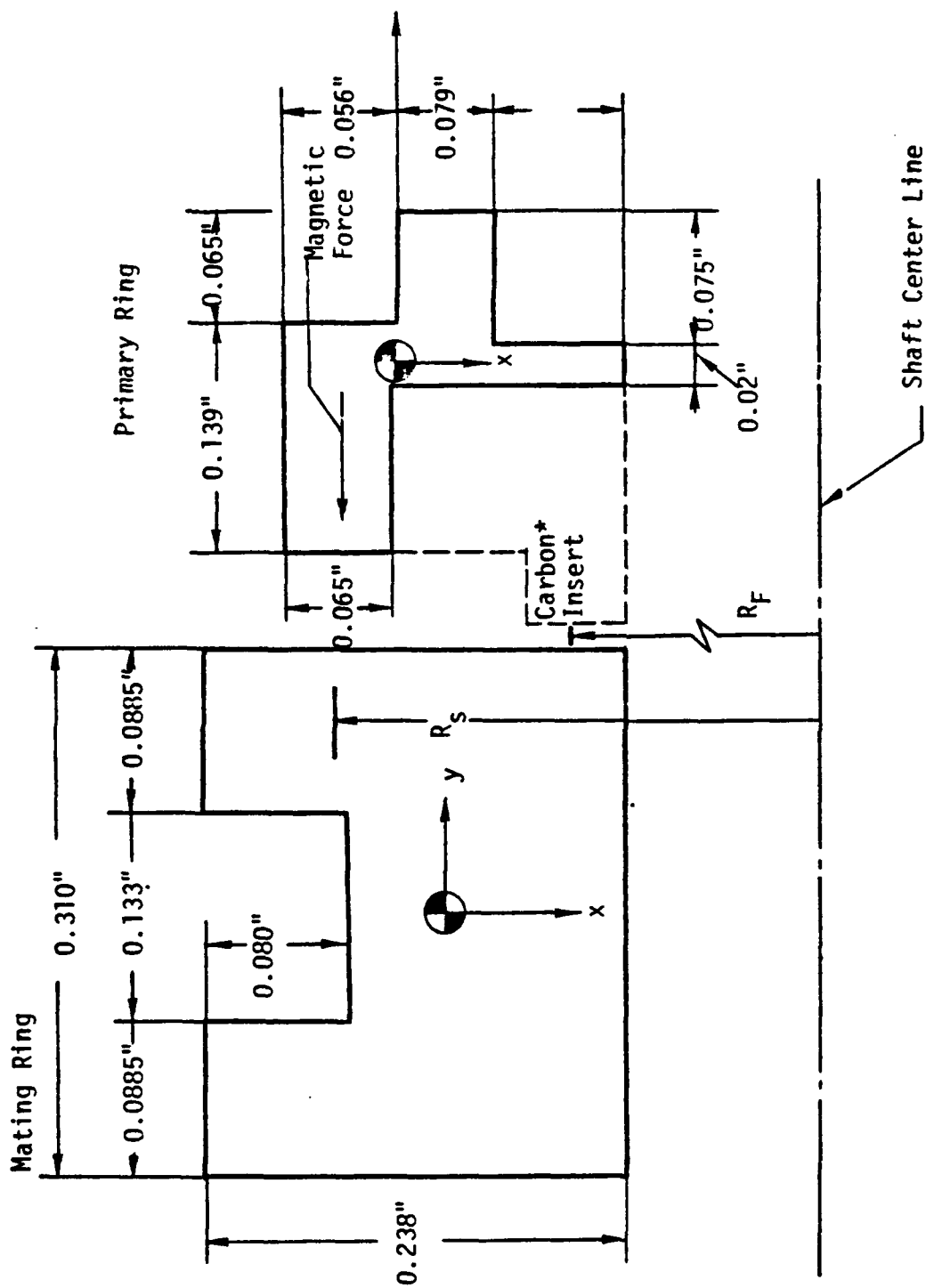
Line 510: RF--Radius to seal face
 Line 520: RS--Radius to support of mating ring
 Line 580: VXP(I)--Applied force in x-direction at node I on
 primary ring
 Line 590: VYP(I)--Applied force in y-direction at node I on
 primary ring
 Line 600: MTP(I)--Applied moment about θ -axis at node I on
 primary ring
 Lines 610-630: VXM(I), VYM(I), MTM(I)--Same as above for mating
 ring
 Line 640: KS(I)--Support stiffness at node I
 Line 650: KF(I)--Face stiffness at node I
 Line 660: DELSI(I)--Initial support profile at node I

 The initial ring face profiles are read from a data file in 6F10.0
 format.

Sample Problem

The modified (reduced section) magnetic face seal shown in Figure
 C-1 is considered as a sample problem. (See Chapter 5 and Ref. [41]).
 The primary ring cross section properties are:

$$\begin{aligned}
 RCP &= 0.6414 \text{ in} \\
 AP &= 0.01603 \text{ in}^2 \\
 JXP &= 4.44 \times 10^{-5} \text{ in}^4 \\
 JYP &= 3.5366 \times 10^{-5} \text{ in}^4 \\
 JXYP &= 2.019 \times 10^{-5} \text{ in}^4 \\
 JTP &= 2.074 \times 10^{-5} \text{ in}^4
 \end{aligned}$$



*Note: The carbon insert is not included in calculation of primary ring properties.

Figure C-1. Reduced Seal Cross Section used for Sample Problem.

The mating ring cross section properties are:

$$RCM = 0.613 \text{ in}$$

$$AM = 0.06314 \text{ in}^2$$

$$JXM = 5.752 \times 10^{-4} \text{ in}^4$$

$$JYM = 2.650 \times 10^{-4} \text{ in}^4$$

$$JXYM = 0.0$$

$$JTM = 4.73 \times 10^{-4} \text{ in}^4$$

The support and face radii are:

$$RS = 0.675 \text{ (the radius at which the magnet acts)}$$

$$RF = 0.553$$

The ring material properties are:

$$EP = 27 \times 10^6 \text{ psi}$$

$$GP = 10.6 \times 10^6 \text{ psi}$$

$$EM = 20 \times 10^6 \text{ psi}$$

$$GP = 8.33 \times 10^6 \text{ psi}$$

The applied loads come from magnetic force of the mating ring acting on the primary ring. The total magnetic force is 1.92 lbs acting in the negative y-direction. It is assumed to act at a .675 radius so it also creates a moment about the θ -axis. The applied forces are:

$$VXP(I) = 0.0$$

$$VYP(I) = -1.92/30$$

$$MTP(I) = -1.92 (0.675 - 0.632^1)/30$$

$$VXM(I) = 0.0$$

$$VYM(I) = 0.0$$

$$MTM(I) = 0.0$$

The stiffness of the carbon insert is the face stiffness:

$$\begin{aligned} KF(I) &= \frac{AE}{L} \\ &= \frac{2\pi(.553)(0.06)(3 \times 10^6)}{0.1156/30} \\ &= 5.4 \times 10^6/30 \end{aligned}$$

The total support stiffness is chosen to be 5.4×10^4 lb/in, 100 times lower to simulate the high conformability of the magnet. Thus

$$KS(I) = 5.4 \times 10^4/30$$

The data file of the initial face profiles is shown in the example.

The program input and output are shown. A plot of the output shows that the gap in the face lessens when the force is applied as expected but does not completely close.

¹ Should have been 0.6441.

AV SAV CAV
0.1576E-03 -0.4757E-05 -0.2041E-04

CONTACTING NODES
10 11 25

PLANE EQUATION COEFFICIENTS
0.4093E-04 0.2340E-04 0.4685E-04

INITIAL FACE DEFLECTIONS

0.9953E-04	0.1102E-03	0.1202E-03	0.1415E-03	0.1290E-03	0.1063E-03
0.7175E-04	0.3325E-04	0.1692E-04	-0.1091E-10	0.0	0.8534E-05
0.2493E-04	0.1897E-04	0.4407E-04	0.1273E-04	0.5426E-05	-0.4919E-05
-0.2116E-04	0.6266E-05	0.9311E-05	0.2495E-04	0.1964E-04	0.7082E-05
0.2638E-10	0.1710E-04	0.3073E-04	0.4515E-04	0.5631E-04	0.7762E-04

INITIAL FACE CONTACT PATTERN

0 0 0 0 0 0 1 1 1 1 1 0 0 0 0 0 1 1 0 0 0 0 1 1 1 1 1 1 1
0
0

FACE CONTACT PATTERN AFTER FORCES ARE APPLIED

0 0 0 0 0 0 0 0 1 1 0 0 0 0 0 0 0 0 1 0 0 0 0 0 1 0 0 0 1 0

FACE DEFLECTIONS AFTER FORCES ARE APPLIED

-0.2002E-04	-0.3338E-04	-0.4259E-04	-0.5970E-04	-0.5526E-04	-0.4431E-04
-0.2353E-04	-0.2131E-06	0.7906E-06	0.4152E-05	-0.4053E-05	-0.1673E-04
-0.3927E-04	-0.3582E-04	-0.6229E-04	-0.3160E-04	-0.2404E-04	-0.1487E-04
0.1073E-05	-0.2541E-04	-0.2684E-04	-0.3974E-04	-0.3015E-04	-0.1144E-04
0.3694E-05	-0.5500E-06	-0.8344E-07	-0.3211E-06	0.1566E-05	-0.7368E-05

FACE FORCES

0.0	0.0	0.0	0.0	0.0	0.0
0.0	-0.3836E-01	0.1437E 00	0.7473E 00	0.0	0.0
0.0	0.0	0.0	0.0	0.0	0.0
0.1932E 00	0.0	0.0	0.0	0.0	0.0
0.6650E 00	0.0	-0.1502E-01	-0.5700E-01	0.2019E 00	0.0

Reprints

FACE CONTACT PATTERN AFTER FORCES ARE APPLIED

0 0 0 0 0 0 0 0 1 1 0 0 0 0 0 0 0 0 1 0 0 0 0 0 1 0 0 0 1 0

FACE DEFLECTIONS AFTER FORCES ARE APPLIED

-0.2049E-04	-0.3391E-04	-0.4310E-04	-0.6033E-04	-0.5590E-04	-0.4493E-04
-0.2409E-04	-0.6033E-06	0.6242E-06	0.4126E-05	-0.3964E-05	-0.1855E-04
-0.3904E-04	-0.3557E-04	-0.6205E-04	-0.3140E-04	-0.2468E-04	-0.1478E-04
0.1111E-05	-0.2539E-04	-0.2685E-04	-0.3970E-04	-0.2021E-04	-0.1154E-04

0.3559E-05 -0.8495E-06 -0.5279E-06 -0.8440E-06 0.1247E-05 -0.7759E-05

FACE FORCES

0.0	0.0	0.0	0.0	0.0	0.0
0.0	0.0	0.1124E 00	0.7426E 00	0.0	0.0
0.0	0.0	0.0	0.0	0.0	0.0
0.2000E 00	0.0	0.0	0.0	0.0	0.0
0.6406E 00	0.0	0.0	0.0	0.2245E 00	0.0

CONVERGENCE IS OK

AVG. GAP	CUBE MEAN
0.2191E 02	0.3376E 02

```

10 C TWO RING DEFLECTION PROGRAM
20 C THIS PROGRAM CALCULATES THE DEFLECTIONS AND FORCES DEVELOPED
30 C IN A TWO RING FACE SEAL SUBJECTED TO OPERATING FORCES. IT
40 C USES THE FINITE ELEMENT METHOD TO MODEL THE SEAL. THE RINGS
50 C ARE MODELED WITH BEAM ELEMENTS AND THE FACE CONTACT IS MODELED
60 C WITH SPRINGS. THE INITIAL AND FINAL FACE DEFLECTIONS AND
70 C FORCES ARE CALCULATED AND THE AVERAGE GAP AND CUBE MEAN OF
80 C THE GAP ARE DETERMINED.
90 C
100 DOUBLE PRECISION KP, KM, AMAT, IMAT, IIMAT, IVMAT, RHA, RHB, BMAT
110 COMMON CVXP(30), CVXM(30), CVYM(30), CVYP(30), CMTM(30), CMTM(30)
120 COMMON CDELF(30), CDELSI(30), CKS(30), CKF(30), KP(12, 12), KM(12, 12)
130 COMMON AMAT(360, 59), IMAT(12, 12), IIMAT(12, 12), IIVMAT(12, 12)
140 COMMON IVMAT(12, 12), BMAT(360), RHA(12), RHB(12)
150 COMMON RCP, RCM, RF, RS
160 REAL*4 MTM, MTP, JXP, JYP, JXYP, JTP, JXM, JYM, JXYM, JTM
170 REAL*4 KS, KF
180 DIMENSION VXP(30), VXM(30), VYM(30), VYP(30), MTM(30), MTP(30)
190 DIMENSION DELSI(30), DELF(30), KS(30), KF(30), DEF(30)
200 DIMENSION III(30), ICOMP(30), ICC(30), ICCOLD(30)
210 DIMENSION UOUT(30), VOUT(30), PHIOUT(30)
220 DOUBLE PRECISION PIE, SUM(12, 12)
230 DIMENSION DELP(30), DELM(30), FORC(30), DELNET(30), DELMTP(30)
240 DIMENSION DELTAP(12), DELTAM(12), FORCE(12)
250 C
260 PIE=4.*DATAN(1.D+00)
270 PIES=PIE
280 C
290 C INPUT NUMBER OF NODES, INITIAL WAVINESS AND RINGS' SECTION
300 C PROPERTIES
310 N=20
320 READ(5, 18) DELP
330 READ(5, 18) DELMTP
340 18 FORMAT(6F10.0)
350 RCP=0.6414
360 AP=0.016035
370 JXP=4.44E-5
380 JYP=3.536E-5
390 JXYP=2.019E-5
400 JTP=2.0738E-5
410 EP=27.E+06
420 GP=10.6E+06
430 RCM=0.613
440 AM=0.06314
450 JXM=5.752E-04
460 JYM=2.65E-04
470 JXYM=1.E-06
480 JTM=4.729E-04
490 EM=20.E+06
500 GM=6.33E+06
510 RF=0.553
520 RS=0.675
530 C
540 C INPUT THE RING LOADS VX, VY, MT HERE OTHER LOADS ARE ASSUMED ZERO.
550 C THESE ARE ACTUAL LOADS AT THE NODES
560 ALOAD=-1.92
570 DO 100 I=1, N
580 VXP(I)=0.
590 VYP(I)=ALOAD/N
600 MTP(I)=-VYP(I)*(.675- .622)
610 VXM(I)=0.
620 VYM(I)=0.
630 MTM(I)=0.

```

```

640         KS(I)=5.4E+04/N
650         KF(I)=5.4E+06/N
660         DELSI(I)=0.
670     100 CONTINUE
680         DT=2.*PIE/N
690         DO 8000 IROT=1,1
700         IMOV=IROT-1
710 C
720 C ELIMINATE FIRST HARMONIC FROM GAP
730     DO 22 I=1,N
740         II=I+IMOV
750         IF(II.GT.N) II=II-N
760     22     DELM(I)=DELMTP(II)
770     DO 20 I=1,N
780     20     DELF(I)=(-DELP(I)-DELM(I))*1.E-06
790         SAV=0.
800         CAV=0.
810         AV=0.
820         NP1=N+1
830         DO 23 I=1,NP1
840         FAC=2.
850         J=I-1
860         IF(I.EQ.1) FAC=1.
870         IF(I.EQ.NP1) FAC=1.
880         II=I
890         IF(I.EQ.NP1) II=1
900         AV=AV+DELF(II)*FAC*.5
910         SAV=SAV+DELF(II)*FAC*SIN(2.*PIES*J/N)
920     25     CAV=CAV+DELF(II)*FAC*COS(2.*PIES*J/N)
930         AV=AV/N
940         SAV=SAV/N
950         CAV=CAV/N
960         WRITE (6,27)
970     27     FORMAT(/,6X,'AV',10X,'SAV',10X,'CAV')
980         WRITE(6,3) AV,SAV,CAV
990         DO 30 I=1,N
1000        J=I-1
1010        DELF(I)=DELF(I)-AV-CAV*COS(2.*PIES*J/N)-SAV*SIN(2.*PIES*J/N)
1020    30 CONTINUE
1030 3     FORMAT(1H,6E12.4)
1040 C TRY PICKING THREE SMALLEST IN THREE SEGMENTS
1050     AMIN=1.E+06
1060     N1=N/3
1070     DO 35 I=1,N1
1080         IF(DELF(I).GT.AMIN) GO TO 35
1090         AMIN=DELF(I)
1100         IA=I
1110    35 CONTINUE
1120     BMIN=1.E+06
1130     N1=N1+1
1140     N2=N1+N/3
1150     DO 36 I=N1,N2
1160         IF(DELF(I).GT.BMIN) GO TO 36
1170         BMIN=DELF(I)
1180         IB=I
1190    36 CONTINUE
1200     N1=N2+1
1210     CMIN=1.E+06
1220     DO 37 I=N1,N
1230         IF(DELF(I).GT.CMIN) GO TO 37
1240         CMIN=DELF(I)
1250         IC=I
1260    37 CONTINUE
1270 4     FORMAT(1H,30I2)
1280     WRITE (6,6)
1290     6     FORMAT(/,1X,'CONTACTING NODES',1X)

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1300      WRITE(6,7) IA, IB, IC
1310      7 FORMAT(3X, 3I3)
1320      TA=(IA-1)*DT
1330      TB=(IB-1)*DT
1340      TC=(IC-1)*DT
1350      DET=(SIN(TA)-SIN(TB))*(COS(TA)-COS(TC))
1360      DET=DET-(COS(TA)-COS(TB))*(SIN(TA)-SIN(TC))
1370      DC2=(DELFI(1B)-DELFI(1A))*(COS(TA)-COS(TC))
1380      DC2=DC2-(DELFI(1C)-DELFI(1A))*(COS(TA)-COS(TB))
1390      C2=DC2/DET
1400      C3=DELFI(1B)-DELFI(1A)-C2*(SIN(TA)-SIN(TB))
1410      C3=C3/(COS(TA)-COS(TB))
1420      C1=-DELFI(1A)-C2*SIN(TA)-C3*COS(TA)
1430      WRITE (6,8)
1440      8 FORMAT(/1X, 'PLANE EQUATION COEFFICIENTS . . .')
1450      WRITE(6,3) C1, C2, C3
1460      DO 39 I=1, N
1470          TT=(I-1)*DT
1480          ICC(I)=0
1490          IF(DELFI(I).LE.0.) ICC(I)=1
1500          DELFI(I)=DELFI(I)+C1+C2*SIN(TT)+C3*COS(TT)
1510          IF(DELFI(I).LT.1.E-06) ICC(I)=1
1520 39      CONTINUE
1530      WRITE (6,40)
1540      40 FORMAT(/1X, 'INITIAL FACE DEFLECTIONS' /)
1550      WRITE(6,3) DELFI
1560      WRITE (6,41)
1570      41 FORMAT(/1X, 'INITIAL FACE CONTACT PATTERN' /)
1580      WRITE(6,4) ICC
1590  C
1600  C REARRANGE SUBSCRIPTS FOR COMPUTATIONAL EFFICIENCY
1610      III(1)=1
1620      III(2)=2
1630      DO 110 I=3, N
1640          AI=I
1650          BI=I/2
1660          AI=AI/2.-BI
1670          IF(AI.GT.0.1) GO TO 120
1680          III(I)=(I+2)/2
1690          GO TO 110
1700      120      III(I)=(2*N+3-I)/2
1710      110 CONTINUE
1720      DO 133 I=1, N
1730          IIN=III(I)
1740      133      ICOMP(IIN)=I
1750      9 FORMAT(1H, 3I5)
1760  C
1770  C COMPUTE ELEMENT STIFFNESSES
1780      CALL AOLCOP(RCP, AP, JXP, JYP, JXYP, JTP, EP, GP, DT, KP)
1790      CALL AOLCOP(RCM, AM, JXM, JYM, JXYM, JTM, EM, GM, CT, KM)
1800      DO 135 I=1, 12
1810          RHA(I)=0.
1820          RHB(I)=0.
1830      DO 135 J=1, 12
1840          IMAT(I, J)=0.
1850          IIMAT(I, J)=0.
1860          IIIMAT(I, J)=0.
1870          IVMAT(I, J)=0.
1880      135 CONTINUE
1890      DO 140 I=1, N
1900          II=III(I)
1910          CDELSI(I)=-DELSI(II)
1920      140      CDELFI(I)=-DELFI(II)
1930      FRAC=0.
1940      NFRAC=8
1950      NITMAX=NFRAC+2

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1960      DO 5000 NIT=1,NITMAX
1970      FRAC=FRAC+1./NFRAC
1980      IF(FRAC.GT.1.0) FRAC=1.
1990      DO 150 I=1,N
2000          II=III(I)
2010          CVXP(I)=FRAC*VXP(II)
2020          CVXM(I)=FRAC*VXM(II)
2030          CVYM(I)=FRAC*VYM(II)
2040          CVYP(I)=FRAC*VYP(II)
2050          CMTM(I)=FRAC*MTM(II)
2060          CMTF(I)=FRAC*MTF(II)
2070          CKS(I)=KS(II)
2080      150      CKF(I)=KF(II)*ICC(II)
2090      NEQ=12*N
2100      MBAND=29
2110  C
2120  C ASSEMBLE GLOBAL STIFFNESS AND FORCE MATRICES
2130      DO 250 I=1,NEQ
2140          BMAT(I)=0.
2150          DO 250 J=1,59
2160      250          AMAT(I,J)=0.
2170          CALL ISUB(1,2)
2180          CALL IVSUB(3,1)
2190          CALL ADD(IMAT,IVMAT,SUM)
2200          CALL IISUB
2210          CALL IIISUB
2220          J1=31
2230          DO 300 I=1,12
2240              BMAT(I)=RHA(I)+RHB(I)
2250              J1=J1-1
2260              J2=J1+11
2270              JJ=0
2280              DO 300 J=J1,J2
2290                  JJ=JJ+1
2300                  AMAT(I,J)=SUM(I,JJ)
2310                  AMAT(I,J+12)=IIMAT(I,JJ)
2320                  IF((J+24).GT.59) GO TO 300
2330                  AMAT(I,J+24)=IIIMAT(I,JJ)
2340      300          CONTINUE
2350          CALL ISUB(2,4)
2360          CALL IVSUB(1,2)
2370          CALL ADD(IMAT,IVMAT,SUM)
2380          J1=31
2390          DO 400 I=13,24
2400              II=I-12
2410              BMAT(I)=RHA(II)+RHB(II)
2420              J1=J1-1
2430              J2=J1+11
2440              JJ=0
2450              DO 400 J=J1,J2
2460                  JJ=JJ+1
2470                  AMAT(I,J)=SUM(II,JJ)
2480                  IF((J+24).GT.59) GO TO 400
2490                  AMAT(I,J+24)=IIMAT(II,JJ)
2500      400          AMAT(I,J-12)=IIIMAT(II,JJ)
2510      NM2=N-2
2520      NODD=1
2530      DO 1000 NN=3,NM2
2540          IST=(NN-1)*12+1
2550          ISTOP=IST+11
2560          IF(NODD.EQ.2) GO TO 1050
2570          CALL ISUB(NN,NN-2)
2580          CALL IVSUB(NN+2,NN)
2590          CALL ADD(IMAT,IVMAT,SUM)
2600          GO TO 1050
2610      1050      CONTINUE

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2620      CALL ISUB(NN, NN+2)
2630      CALL IVSUB(NN-2, NN)
2640      CALL ADD(IMAT, IVMAT, SUM)
2650 1060  J1=31
2660      II=0
2670      DO 1070 I=IST, ISTOP
2680          II=II+1
2690          BMAT(I)=RHA(II)+RHB(II)
2700          J1=J1-1
2710          J2=J1+11
2720          JJ=0
2730          DO 1070 J=J1, J2
2740              JJ=JJ+1
2750              AMAT(I, J)=SUM(II, JJ)
2760              GO TO (1080, 1090), NODD
2770 1080  IF((J-24).LT.1) GO TO 1081
2780          AMAT(I, J-24)=IIMAT(II, JJ)
2790 1081  IF((J+24).GT.59) GO TO 1070
2800          AMAT(I, J+24)=IIMAT(II, JJ)
2810          GO TO 1070
2820 1090  IF((J-24).LT.1) GO TO 1091
2830          AMAT(I, J-24)=IIMAT(II, JJ)
2840 1091  IF((J+24).GT.59) GO TO 1070
2850          AMAT(I, J+24)=IIMAT(II, JJ)
2860 1070  CONTINUE
2870          IF(NODD.EQ.1) GO TO 1095
2880          NODD=1
2890          GO TO 1080
2900 1095  NODD=2
2910 1080  CONTINUE
2920      NM1=N-1
2930      CALL ISUB(NM1, NM1-2)
2940      CALL IVSUB(N, N-1)
2950      CALL ADD(IMAT, IVMAT, SUM)
2960      IST=NM2+12+1
2970      ISTOP=IST+11
2980      J1=31
2990      II=0
3000      DO 500 I=IST, ISTOP
3010          II=II+1
3020          BMAT(I)=RHA(II)+RHB(II)
3030          J1=J1-1
3040          J2=J1+11
3050          JJ=0
3060          DO 500 J=J1, J2
3070              JJ=JJ+1
3080              AMAT(I, J)=SUM(II, JJ)
3090              IF((J-24).LT.1) GO TO 500
3100              AMAT(I, J-24)=IIMAT(II, JJ)
3110 500   AMAT(I, J+12)=IIMAT(II, JJ)
3120      CALL ISUB(N, N-1)
3130      CALL IVSUB(N-2, N)
3140      CALL ADD(IMAT, IVMAT, SUM)
3150      IST=NM1+12+1
3160      ISTOP=IST+11
3170      J1=31
3180      II=0
3190      DO 600 I=IST, ISTOP
3200          II=II+1
3210          BMAT(I)=RHA(II)+RHB(II)
3220          J1=J1-1
3230          J2=J1+11
3240          JJ=0
3250          DO 600 J=J1, J2
3260              JJ=JJ+1
3270              AMAT(I, J)=SUM(II, JJ)

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3280             IF((J-24).LT.1) GO TO 600
3290             AMAT(I,J-24)=IIMAT(II,JJ)
3300 600         AMAT(I,J-12)=IIMAT(II,JJ)
3310             DO 700 NN=1,N
3320                 NC=(NN-1)*12+1
3330                 BMAT(NC)=BMAT(NC)+CVXP(NN)
3340                 BMAT(NC+1)=BMAT(NC+1)+CVYP(NN)
3350                 BMAT(NC+5)=BMAT(NC+5)+CMTN(NN)
3360                 BMAT(NC+6)=BMAT(NC+6)+CVXM(NN)
3370                 BMAT(NC+7)=BMAT(NC+7)+CVYM(NN)
3380 700         BMAT(NC+11)=BMAT(NC+11)+CMTM(NN)
3390 C
3400 C ENFORCE ZERO DISPLACEMENT BOUNDARY CONDITIONS:
3410 C UP1, UM1, WP1, WM1, WPN, WMN=0
3420             II=12*NM1
3430             DO 750 J=1,59
3440                 AMAT(1,J)=0.
3450                 AMAT(7,J)=0.
3460                 AMAT(3,J)=0.
3470                 AMAT(9,J)=0.
3480                 AMAT(II+3,J)=0.
3490 750         AMAT(II+9,J)=0.
3500             AMAT(1,30)=1.
3510             AMAT(7,30)=1.
3520             AMAT(3,30)=1.
3530             AMAT(9,30)=1.
3540             AMAT(II+3,30)=1.
3550             AMAT(II+9,30)=1.
3560             BMAT(1)=0.
3570             BMAT(3)=0.
3580             BMAT(7)=0.
3590             BMAT(9)=0.
3600             BMAT(II+3)=0.
3610             BMAT(II+9)=0.
3620 C
3630 C SOLVE FOR DEFLECTIONS
3640             CALL GAUSS(AMAT,BMAT,MBAND,NEQ)
3650 C
3660 C CALCULATE AND PRINT DEFLECTIONS AND FORCES
3670             ICCSUM=0
3680             DO 900 I=1,N
3690                 II=(ICOMP(I)-1)*12
3700                 DEF(I)=BMAT(II+8)-BMAT(II+2)-BMAT(II+12)*(RF-RCM)
3710                 DEF(I)=DEF(I)+BMAT(II+6)*(RF-RCP)
3720                 FORC(I)=(DEF(I)-DELF(I))*KF(I)*ICC(I)
3730                 DELNET(I)=DEF(I)-DELF(I)
3740                 ICCOLD(I)=ICC(I)
3750                 ICC(I)=0
3760                 IF(DEF(I).GE.DELF(I)) ICC(I)=1
3770                 ICCSUM=ICCSUM+(ICC(I)-ICCOLD(I))*2
3780 900         CONTINUE
3790             IF(FRAC.NE.1.) GO TO 5000
3800             WRITE(6,905)
3810 905         FORMAT(//1X,'FACE CONTACT PATTERN AFTER FORCES ARE APPLIED'//)
3820             WRITE(6,4) ICC
3830             WRITE(6,906)
3840 906         FORMAT(//1X,'FACE DEFLECTIONS AFTER FORCES ARE APPLIED'//)
3850             WRITE(6,3) DELNET
3860             WRITE(6,907)
3870 907         FORMAT(//1X,'FACE FORCES'//)
3880             WRITE(6,3) FORC
3890             IF(ICCSUM.EQ.0) GO TO 6000
3900 5000     CONTINUE
3910             GO TO 6001
3920 6000     WRITE(6,17)
3930 17       FORMAT(//1X,'CONVERGENCE IS OK')

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3940 C
3950 C COMPUTE AVERAGE GAP AND CUBE MEAN OF THE GAP
3960 8001 HAV=0.
3970 HAV3=0.
3980 DO 8002 I=1, N
3990 IF (DELNET(I).GT.0.) GO TO 8002
4000 HAV=HAV+DELNET(I)
4010 HAV3=HAV3+DELNET(I)**3
4020 8002 CONTINUE
4030 HAV=HAV*1. E+06/N
4040 HAV3=(HAV3/N)**0.333333333333*1. E+06
4050 WRITE (6,8003)
4060 8003 FORMAT(/,3X,'AVG. GAP CUBE MEAN')
4070 WRITE(6,8004) HAV, HAV3
4080 8004 FORMAT(1X, E12. 4, E12. 4)
4090 8000 CONTINUE
4100 STOP
4110 END
4120 C SUBROUTINES USED BY THE PROGRAM
4130 SUBROUTINE ISUB(IA, IB)
4140 DOUBLE PRECISION KP, KM, AMAT, IMAT, IIMAT, IIMAT, IVMAT, RHA, RHB, BMAT
4150 COMMON CVXP(30), CVXM(30), CVYM(30), CVYP(30), CMTM(30), CMTM(30)
4160 COMMON CDELF(30), CDELSI(30), CKS(30), CKF(30), KP(12, 12), KM(12, 12)
4170 COMMON AMAT(360, 59), IMAT(12, 12), IIMAT(12, 12), IIMAT(12, 12)
4180 COMMON IVMAT(12, 12), BMAT(360), RHA(12), RHB(12)
4190 COMMON RCP, RCM, RF, RS
4200 DO 10 I=1, 6
4210 DO 10 J=1, 6
4220 10 IMAT(I, J)=KP(I, J)
4230 DO 20 I=7, 12
4240 DO 20 J=7, 12
4250 II=I-6
4260 JJ=J-6
4270 20 IMAT(I, J)=KM(II, JJ)
4280 IMAT(2, 2)=IMAT(2, 2)+CKF(IA)*.5
4290 IMAT(2, 6)=IMAT(2, 6)-CKF(IA)*.5*(RF-RCP)
4300 IMAT(2, 8)=-CKF(IA)*.5
4310 IMAT(2, 12)=CKF(IA)*.5*(RF-RCM)
4320 IMAT(6, 2)=IMAT(6, 2)-CKF(IA)*.5*(RF-RCP)
4330 IMAT(6, 6)=IMAT(6, 6)+CKF(IA)*.5*(RF-RCP)**2
4340 IMAT(6, 8)=CKF(IA)*.5*(RF-RCP)
4350 IMAT(6, 12)=-CKF(IA)*.5*(RF-RCM)*(RF-RCP)
4360 IMAT(8, 2)=-CKF(IA)*.5
4370 IMAT(8, 6)=CKF(IA)*.5*(RF-RCP)
4380 IMAT(8, 8)=IMAT(8, 8)+CKF(IA)*.5+CKS(IA)*.5
4390 IMAT(8, 12)=IMAT(8, 12)-CKF(IA)*.5*(RF-RCM)+CKS(IA)*.5*(RCM-RS)
4400 IMAT(12, 2)=CKF(IA)*.5*(RF-RCM)
4410 IMAT(12, 6)=-CKF(IA)*.5*(RF-RCM)*(RF-RCP)
4420 IMAT(12, 8)=IMAT(12, 8)-CKF(IA)*.5*(RF-RCM)+CKS(IA)*.5*(RCM-RS)
4430 IMAT(12, 12)=IMAT(12, 12)+CKF(IA)*.5*(RF-RCM)**2+CKS(IA)*.5*(RCM-RS)
4440 ***2
4450 RHA(2)=CKF(IA)*.5+CDELF(IA)
4460 RHA(6)=-CKF(IA)*.5+CDELF(IA)*(RF-RCP)
4470 RHA(8)=-CKF(IA)*.5+CDELF(IA)+CKS(IA)*.5+CDELSI(IA)
4480 RHA(12)=CKF(IA)*.5+CDELF(IA)*(RF-RCM)+CKS(IA)*.5+CDELSI(IA)*(RCM-
4490 *RS)
4500 RETURN
4510 END
4520 C
4530 C
4540 SUBROUTINE IVSUB(IA, IB)
4550 DOUBLE PRECISION KP, KM, AMAT, IMAT, IIMAT, IIMAT, IVMAT, RHA, RHB, BMAT
4560 COMMON CVXP(30), CVXM(30), CVYM(30), CVYP(30), CMTM(30), CMTM(30)
4570 COMMON CDELF(30), CDELSI(30), CKS(30), CKF(30), KP(12, 12), KM(12, 12)
4580 COMMON AMAT(360, 59), IMAT(12, 12), IIMAT(12, 12), IIMAT(12, 12)
4590 COMMON IVMAT(12, 12), BMAT(360), RHA(12), RHB(12)

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4600      COMMON RCP, RCM, RF, RS
4610      DO 10 I=1, 6
4620          DO 10 J=1, 6
4630              II=I+6
4640              JJ=J+6
4650          10      IVMAT(I, J)=KP(II, JJ)
4660      DO 20 I=7, 12
4670          DO 20 J=7, 12
4680      20      IVMAT(I, J)=KM(I, J)
4690      IVMAT(2, 2)=IVMAT(2, 2)+CKF(IB)*. 5
4700      IVMAT(2, 6)=IVMAT(2, 6)-CKF(IB)*. 5*(RF-RCP)
4710      IVMAT(2, 8)=-CKF(IB)*. 5
4720      IVMAT(2, 12)=CKF(IB)*. 5*(RF-RCM)
4730      IVMAT(6, 2)=-CKF(IB)*. 5*(RF-RCP)+IVMAT(6, 2)
4740      IVMAT(6, 6)=IVMAT(6, 6)+CKF(IB)*. 5*(RF-RCP)**2
4750      IVMAT(6, 8)=CKF(IB)*. 5*(RF-RCP)
4760      IVMAT(6, 12)=-CKF(IB)*. 5*(RF-RCM)*(RF-RCP)
4770      IVMAT(8, 2)=-CKF(IB)*. 5
4780      IVMAT(8, 6)=CKF(IB)*. 5*(RF-RCP)
4790      IVMAT(8, 8)=IVMAT(8, 8)+CKF(IB)*. 5+CKS(IB)*. 5
4800      IVMAT(8, 12)=IVMAT(8, 12)-CKF(IB)*. 5*(RF-RCM)+CKS(IB)*. 5*(RCM-RS)
4810      IVMAT(12, 2)=CKF(IB)*. 5*(RF-RCM)
4820      IVMAT(12, 6)=-CKF(IB)*. 5*(RF-RCM)*(RF-RCP)
4830      IVMAT(12, 8)=-CKF(IB)*. 5*(RF-RCM)+CKS(IB)*. 5*(RCM-RS)+IVMAT(12, 8)
4840      IVMAT(12, 12)=IVMAT(12, 12)+CKF(IB)*. 5*(RF-RCM)**2+CKS(IB)*. 5*(RCM-R
4850      *S)**2
4860      RHB(2)=CKF(IB)*. 5*CDELFI(IB)
4870      RHB(6)=-CKF(IB)*. 5*CDELFI(IB)*(RF-RCP)
4880      RHB(8)=-CKF(IB)*. 5*CDELFI(IB)+CKS(IB)*. 5*CDELSI(IB)
4890      RHB(12)=CKF(IB)*. 5*CDELFI(IB)*(RF-RCM)+CKS(IB)*. 5*CDELSI(IB)*(RCM-R
4900      *RS)
4910      RETURN
4920      END
4930 C
4940 C
4950      SUBROUTINE IIISUB
4960      DOUBLE PRECISION KP, KM, AMAT, IMAT, IIMAT, IIIMAT, IVMAT, RHA, RHB, BMAT
4970      COMMON CVXP(30), CVXM(30), CVYM(30), CVYP(30), CMTM(30), CMTF(30)
4980      COMMON CDELFI(30), CDELSI(30), CKS(30), CKF(30), KP(12, 12), KM(12, 12)
4990      COMMON AMAT(360, 59), IMAT(12, 12), IIMAT(12, 12), IIIMAT(12, 12)
5000      COMMON IVMAT(12, 12), BMAT(360), RHA(12), RHB(12)
5010      COMMON RCP, RCM, RF, RS
5020      DO 10 I=1, 6
5030          DO 10 J=1, 6
5040              JJ=J+6
5050          10      IIMAT(I, J)=KP(I, JJ)
5060      DO 20 I=7, 12
5070          DO 20 J=7, 12
5080              II=I-6
5090          20      IIMAT(I, J)=KM(II, J)
5100      RETURN
5110      END
5120 C
5130 C
5140      SUBROUTINE IIISUB
5150      DOUBLE PRECISION KP, KM, AMAT, IMAT, IIMAT, IIIMAT, IVMAT, RHA, RHB, BMAT
5160      COMMON CVXP(30), CVXM(30), CVYM(30), CVYP(30), CMTM(30), CMTF(30)
5170      COMMON CDELFI(30), CDELSI(30), CKS(30), CKF(30), KP(12, 12), KM(12, 12)
5180      COMMON AMAT(360, 59), IMAT(12, 12), IIMAT(12, 12), IIIMAT(12, 12)
5190      COMMON IVMAT(12, 12), BMAT(360), RHA(12), RHB(12)
5200      COMMON RCP, RCM, RF, RS
5210      DO 10 I=1, 6
5220          DO 10 J=1, 6
5230              II=I+6
5240          10      IIIMAT(I, J)=KP(II, J)
5250      DO 20 I=7, 12

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5260      DO 20 J=7,12
5270          JJ=J-6
5280      20      IIIMAT(I,J)=KM(I,JJ)
5290      RETURN
5300      END
5310 C
5320 C
5330      SUBROUTINE ADD(MATA,MATB,MATSUM)
5340      DOUBLE PRECISION MATA(12,12),MATB(12,12),MATSUM(12,12)
5350      DO 10 I=1,12
5360          DO 10 J=1,12
5370      10      MATSUM(I,J)=MATA(I,J)+MATB(I,J)
5380      RETURN
5390      END

5400 C
5410 C AOLCOP - ELEMENT STIFFNESS MATRIX FOR COUPLED RING
5420      SUBROUTINE AOLCOP(RC,AREA,JX,JY,JXY,JT,E,G,DT,K)
5430      REAL*4 JX,JY,JXY,JT
5440      DOUBLE PRECISION A(12,12),K(12,12),WORK(12),DET(2),D(12,12)
5450      DOUBLE PRECISION T(2),PIE,R,AA,JYOR2,JXYOJX,JXYOJY,P,Q,SS
5460      DOUBLE PRECISION U,V,S,C,TH,F
5470      DIMENSION IPVT(12)
5480      PIE=4.*DATAN(1.D+00)
5490      T(1)=1.
5500      T(2)=T(1)+DT
5510      R=RC
5520      AA=E+JX/G/JT
5530      JYOR2=JY/AREA/R/R
5540      JXYOJX=JXY/JX
5550      JXYOJY=JXY/JY
5560      P=1./JXYOJX
5570      Q=0.5*(1.+1./AA)/JXYOJX
5580      SS=(1.+JYOR2)/JXYOJX-JYOR2+JXYOJY
5590      U=1./JXYOJY-JXYOJX
5600      V=(1.+1./AA)/JXYOJY-JXYOJX
5610      DO 10 I=1,12
5620          DO 10 J=1,12
5630              A(I,J)=0.
5640              K(I,J)=0.
5650      10      D(I,J)=0.
5660      F=1.
5670      DO 20 II=1,2
5680          TH=T(II)
5690          C=DCOS(TH)
5700          S=DSIN(TH)
5710          I=1+(II-1)*6
5720          A(I,3)=-Q*TH*C
5730          A(I,4)=Q*TH*S
5740          A(I,5)=-P*TH*S
5750          A(I,6)=-P*TH*C
5760          A(I,7)=P
5770          A(I,8)=-Q*TH*C
5780          A(I,9)=Q*TH*S
5790          A(I,10)=P*S
5800          A(I,11)=P*C
5810          I=I+1
5820          A(I,1)=1.
5830          A(I,2)=TH
5840          A(I,3)=S
5850          A(I,4)=C
5860          A(I,5)=TH*S
5870          A(I,6)=TH*C
5880          I=I+1
5890          A(I,3)=-Q*(C+TH*S)
5900          A(I,4)=Q*(S-TH*C)

```

```

5910      A(I, 5)=-P*(S-TH*C)
5920      A(I, 6)=-P*(C+TH*S)
5930      A(I, 7)=SS*TH
5940      A(I, 8)=-Q*(C+TH*S)
5950      A(I, 9)=Q*(S-TH*C)
5960      A(I, 11)=P*S
5970      A(I, 10)=-P*C
5980      A(I, 12)=1.
5990      I=I+1
6000      A(I, 2)=1./R
6010      A(I, 3)=C/R
6020      A(I, 4)=-S/R
6030      A(I, 5)=(TH*C+S)/R
6040      A(I, 6)=(-TH*S+C)/R
6050      I=I+1
6060      A(I, 3)=-2.*Q*C/R
6070      A(I, 4)=2.*Q*S/R
6080      A(I, 5)=-2.*P*S/R
6090      A(I, 6)=-2.*P*C/R
6100      A(I, 7)=SS*TH/R
6110      A(I, 8)=-2.*Q*C/R
6120      A(I, 9)=2.*Q*S/R
6130      A(I, 12)=1./R
6140      I=I+1
6150      A(I, 5)=-TH*S/R
6160      A(I, 6)=-TH*C/R
6170      A(I, 7)=1./R
6180      A(I, 8)=S/R
6190      A(I, 9)=C/R
6200      I=1+(II-1)*6
6210      D(I, 3)=V*C*F/R
6220      D(I, 4)=-V*S*F/R
6230      D(I, 5)=2.*U*S*F/R
6240      D(I, 6)=2.*U*C*F/R
6250      D(I, 8)=V*C*F/R
6260      D(I, 9)=-V*S*F/R
6270      I=I+1
6280      D(I, 2)=-1.*F/R/AA
6290      I=I+1
6300      D(I, 3)=V*S*F/R
6310      D(I, 4)=V*C*F/R
6320      D(I, 5)=-2.*U*C*F/R
6330      D(I, 6)=2.*U*S*F/R
6340      D(I, 8)=V*S*F/R
6350      D(I, 9)=V*C*F/R
6360      I=I+1
6370      D(I, 3)=-S*F/AA
6380      D(I, 4)=-C*F/AA
6390      D(I, 8)=-S*F/AA
6400      D(I, 9)=-C*F/AA
6410      I=I+1
6420      D(I, 3)=-V*S*F
6430      D(I, 4)=-V*C*F
6440      D(I, 5)=2.*U*C*F
6450      D(I, 6)=-2.*U*S*F
6460      D(I, 7)=-U*F
6470      D(I, 8)=-V*S*F
6480      D(I, 9)=-V*C*F
6490      I=I+1
6500      D(I, 2)=-F/AA
6510      D(I, 3)=-C*F/AA
6520      D(I, 4)=S*F/AA
6530      D(I, 8)=-C*F/AA
6540      D(I, 9)=S*F/AA
6550      F=-1.
6560      20 CONTINUE

```



```

6570      CALL DGEFA(A,12,12,IPVT,INFO)
6580      WRITE(6,1) INFO
6590      1 FORMAT(1H ,15)
6600      JOB=1
6610      CALL DGED1(A,12,12,IPVT,DET,WORK,JOB)
6620      DO 30 I=1,12
6630          DO 30 J=1,12
6640              DO 40 JJ=1,12
6650                  40      K(I,J)=K(I,J)+D(I,JJ)*A(JJ,J)
6660                      K(I,J)=K(I,J)*E*JX/R**2
6670      30 CONTINUE
6680      2  FORMAT(1H ,6E12.4)
6690      RETURN
6700      END
6710 C
6720 C
6730      SUBROUTINE DAXPY(N,DA,DX,INCX,DY,INCY)
6740      DOUBLE PRECISION DX(1),DY(1),DA
6750      INTEGER I,INCX,INCY,IXIY,M,MP1,N
6760      IF(N.LE.0)RETURN
6770      IF(DA.EQ.0.0D0) RETURN
6780      IF(INCX.EQ.1.AND.INCY.EQ.1)GO TO 20
6790      IX = 1
6800      IY = 1
6810      IF(INCX.LT.0)IX = (-N+1)*INCX + 1
6820      IF(INCY.LT.0)IY = (-N+1)*INCY + 1
6830      DO 10 I = 1,N
6840          DY(IY) = DY(IY) + DA*DX(IX)
6850          IX = IX + INCX
6860          IY = IY + INCY
6870      10 CONTINUE
6880      RETURN
6890      20 M = MOD(N,4)
6900      IF(M.EQ.0) GO TO 40
6910      DO 30 I = 1,M
6920          DY(I) = DY(I) + DA*DX(I)
6930      30 CONTINUE
6940      IF(N.LT.4) RETURN
6950      40 MP1 = M + 1
6960      DO 50 I = MP1,N,4
6970          DY(I) = DY(I) + DA*DX(I)
6980          DY(I + 1) = DY(I + 1) + DA*DX(I + 1)
6990          DY(I + 2) = DY(I + 2) + DA*DX(I + 2)
7000          DY(I + 3) = DY(I + 3) + DA*DX(I + 3)
7010      50 CONTINUE
7020      RETURN
7030      END
7040 C
7050 C
7060      INTEGER FUNCTION IDAMAX(N,DX,INCX)
7070      DOUBLE PRECISION DX(1),DMAX
7080      INTEGER I,INCX,IX,N
7090      IDAMAX = 0
7100      IF(N.LT.1) RETURN
7110      IDAMAX = 1
7120      IF(N.EQ.1)RETURN
7130      IF(INCX.EQ.1)GO TO 20
7140      IX = 1
7150      DMAX = DABS(DX(1))
7160      IX = IX + INCX
7170      DO 10 I = 2,N
7180          IF(DABS(DX(IX)).LE.DMAX) GO TO 5
7190          IDAMAX = I
7200          DMAX = DABS(DX(IX))
7210      5  IX = IX + INCX
7220      10 CONTINUE

```

```

7230      RETURN
7240      20 DMAX = DABS(DX(1))
7250      DO 30 I = 2, N
7260          IF(DABS(DX(I)) .LE. DMAX) GO TO 20
7270          IDAMAX = I
7280          DMAX = DABS(DX(I))
7290      30 CONTINUE
7300      RETURN
7310      END
7320 C
7330 C
7340      SUBROUTINE DSCAL(N, DA, DX, INCX)
7350      DOUBLE PRECISION DA, DX(1)
7360      INTEGER I, INCX, M, MP1, N, NINCX
7370      IF(N .LE. 0) RETURN
7380      IF(INCX .EQ. 1) GO TO 20
7390      NINCX = N*INCX
7400      DO 10 I = 1, NINCX, INCX
7410          DX(I) = DA*DX(I)
7420      10 CONTINUE
7430      RETURN
7440      20 M = MOD(N, 5)
7450      IF( M .EQ. 0 ) GO TO 40
7460      DO 30 I = 1, M
7470          DX(I) = DA*DX(I)
7480      30 CONTINUE
7490      IF( N .LT. 5 ) RETURN
7500      40 MP1 = M + 1
7510      DO 50 I = MP1, N, 5
7520          DX(I) = DA*DX(I)
7530          DX(I + 1) = DA*DX(I + 1)
7540          DX(I + 2) = DA*DX(I + 2)
7550          DX(I + 3) = DA*DX(I + 3)
7560          DX(I + 4) = DA*DX(I + 4)
7570      50 CONTINUE
7580      RETURN
7590      END
7600 C
7610 C
7620      SUBROUTINE DSWAP (N, DX, INCX, DY, INCY)
7630      DOUBLE PRECISION DX(1), DY(1), DTEMP
7640      INTEGER I, INCX, INCY, IX, IY, M, MP1, N
7650      IF(N .LE. 0) RETURN
7660      IF(INCX .EQ. 1. AND. INCY .EQ. 1) GO TO 20
7670      IX = 1
7680      IY = 1
7690      IF(INCX .LT. 0) IX = (-N+1)*INCX + 1
7700      IF(INCY .LT. 0) IY = (-N+1)*INCY + 1
7710      DO 10 I = 1, N
7720          DTEMP = DX(IX)
7730          DX(IX) = DY(IY)
7740          DY(IY) = DTEMP
7750          IX = IX + INCX
7760          IY = IY + INCY
7770      10 CONTINUE
7780      RETURN
7790      20 M = MOD(N, 3)
7800      IF( M .EQ. 0 ) GO TO 40
7810      DO 30 I = 1, M
7820          DTEMP = DX(I)
7830          DX(I) = DY(I)
7840          DY(I) = DTEMP
7850      30 CONTINUE
7860      IF( N .LT. 3 ) RETURN
7870      40 MP1 = M + 1
7880      DO 50 I = MP1, N, 3

```

```

7890      DTEMP = DX(I)
7900      DX(I) = DY(I)
7910      DY(I) = DTEMP
7920      DTEMP = DX(I + 1)
7930      DX(I + 1) = DY(I + 1)
7940      DY(I + 1) = DTEMP
7950      DTEMP = DX(I + 2)
7960      DX(I + 2) = DY(I + 2)
7970      DY(I + 2) = DTEMP
7980 50 CONTINUE
7990      RETURN
8000      END
8010 C
8020 C
8030      SUBROUTINE DGEFA(A, LDA, N, IPVT, INFO)
8040      INTEGER LDA, N, IPVT(1), INFO
8050      DOUBLE PRECISION A(LDA, 1)
8060      DOUBLE PRECISION T
8070      INTEGER IDAMAX, J, K, KP1, L, NM1
8080      INFO = 0
8090      NM1 = N - 1
8100      IF (NM1 .LT. 1) GO TO 70
8110      DO 60 K = 1, NM1
8120          KP1 = K + 1
8130          L = IDAMAX(N-K+1, A(K, K), 1) + K - 1
8140          IPVT(K) = L
8150          IF (A(L, K) .EQ. 0.0D0) GO TO 40
8160          IF (L .EQ. K) GO TO 10
8170          T = A(L, K)
8180          A(L, K) = A(K, K)
8190          A(K, K) = T
8200 10      CONTINUE
8210          T = -1.0D0/A(K, K)
8220          CALL DSCAL(N-K, T, A(K+1, K), 1)
8230          DO 30 J = KP1, N
8240              T = A(L, J)
8250              IF (L .EQ. K) GO TO 20
8260              A(L, J) = A(K, J)
8270              A(K, J) = T
8280 20      CONTINUE
8290          CALL DAXPY(N-K, T, A(K+1, K), 1, A(K+1, J), 1)
8300 30      CONTINUE
8310          GO TO 50
8320 40      CONTINUE
8330          INFO = K
8340 50      CONTINUE
8350 60 CONTINUE
8360 70 CONTINUE
8370      IPVT(N) = N
8380      IF (A(N, N) .EQ. 0.0D0) INFO = N
8390      RETURN
8400      END
8410 C
8420 C
8430      SUBROUTINE DGEDI(A, LDA, N, IPVT, DET, WORK, JOB)
8440      INTEGER LDA, N, IPVT(1), JOB
8450      DOUBLE PRECISION A(LDA, 1), DET(2), WORK(1)
8460      DOUBLE PRECISION T
8470      DOUBLE PRECISION TEN
8480      INTEGER I, J, K, KB, KP1, L, NM1
8490      IF (JOB/10 .EQ. 0) GO TO 70
8500      DET(1) = 1.0D0
8510      DET(2) = 0.0D0
8520      TEN = 10.0D0
8530      DO 50 I = 1, N
8540          IF (IPVT(I) .NE. I) DET(1) = -DET(1)

```

```

8550      DET(1) = A(I,1)*DET(1)
8560      IF (DET(1) .EQ. 0.000) GO TO 60
8570  10    IF (DABS(DET(1)) .GE. 1.000) GO TO 20
8580      DET(1) = TEN*DET(1)
8590      DET(2) = DET(2) - 1.000
8600      GO TO 10
8610  20    CONTINUE
8620  30    IF (DABS(DET(1)) .LT. TEN) GO TO 40
8630      DET(1) = DET(1)/TEN
8640      DET(2) = DET(2) + 1.000
8650      GO TO 30
8650  40    CONTINUE
8670  50    CONTINUE
8680  60    CONTINUE
8690  70    CONTINUE
8700      IF (MOD(JOB,10) .EQ. 0) GO TO 150
8710      DO 100 K = 1, N
8720          A(K,K) = 1.000/A(K,K)
8730          T = -A(K,K)
8740          CALL DSCAL(K-1,T,A(1,K),1)
8750          KP1 = K + 1
8760          IF (N .LT. KP1) GO TO 90
8770          DO 80 J = KP1, N
8780              T = A(K,J)
8790              A(K,J) = 0.000
8800              CALL DAXPY(K,T,A(1,K),1,A(1,J),1)
8810  80      CONTINUE
8820  90      CONTINUE
8830  100     CONTINUE
8840          NM1 = N - 1
8850          IF (NM1 .LT. 1) GO TO 140
8860          DO 130 KB = 1, NM1
8870              K = N - KB
8880              KP1 = K + 1
8890              DO 110 I = KP1, N
8900                  WORK(I) = A(I,K)
8910                  A(I,K) = 0.000
8920  110     CONTINUE
8930              DO 120 J = KP1, N
8940                  T = WORK(J)
8950                  CALL DAXPY(N,T,A(1,J),1,A(1,K),1)
8960  120     CONTINUE
8970              L = IPVT(K)
8980              IF (L .NE. K) CALL DSWAP(N,A(1,K),1,A(1,L),1)
8990  130     CONTINUE
9000  140     CONTINUE
9010  150     CONTINUE
9020      RETURN
9030      END
9040 C
9050 C
9060      SUBROUTINE GAUSS(A,B,MBAND,NEQ)
9070 C THIS SUBPROGRAM PERFORMS A GAUSS ELIMINATION
9080 C ON A NON SYMMETRICAL BANDED MATRIX
9090      DOUBLE PRECISION A(6200),B(100),C,S
9100      NQ1=NEQ-1
9110      DO 10 I=1,NQ1
9120          I1=I+1
9130          I2=I+MBAND
9140          DO 20 II=I1,I2
9150              J=I
9160              K=(J-II+MBAND)*NEQ+II
9170              IF(A(K) .EQ. 0.) GO TO 20
9180              KK=(J-I+MBAND)*NEQ+I
9190              C=A(K)/A(KK)
9200              J1=J

```

```

9210          J2=J+MBAND
9220          IF(J2.GT.NEQ) J2=NEQ
9230          DO 40 JJ=J1,J2
9240              KKK=(JJ-II+MBAND)*NEQ+II
9250              KKKK=(JJ-I+MBAND)*NEQ+I
9260          40      A(KKK)=A(KKK)-C*A(KKKK)
9270              B(II)=B(II)-C*B(I)
9280          20      CONTINUE
9290          10      CONTINUE
9300              K=MBAND*NEQ+NEQ
9310              B(NEQ)=B(NEQ)/A(K)
9320          DO 50 II=1,NQ1
9330              I=NEQ-II
9340              J1=I+1
9350              J2=I+MBAND
9360              S=0.
9370              IF(J2.GT.NEQ) J2=NEQ
9380              DO 60 J=J1,J2
9390                  K=(J-I+MBAND)*NEQ+I
9400          60      S=S+A(K)*B(J)
9410              KK=MBAND*NEQ+I
9420          50      B(I)=(B(I)-S)/A(KK)
9430          RETURN
9440          END

```

Face Profiles

10	2.27	-13.95	-27.26	-24.34	-19.47	-3.57
20	18.17	34.40	31.48	34.07	29.21	12.98
30	-15.90	-8.44	-38.29	-11.68	-3.57	3.57
40	12.01	-1.30	5.52	-9.41	-7.14	-4.54
50	<u>5.84</u>	9.74	14.28	16.23	23.69	18.17
60	-151.22	-149.27	-144.73	-162.90	-160.30	-161.60
70	-159.01	-149.27	-144.08	-144.73	-155.11	-162.25
80	-163.55	-177.18	-182.37	-184.97	-190.16	-188.21
90	-178.48	-187.56	-189.51	-179.77	-164.20	-140.18
100	-128.50	-134.34	-137.59	-140.18	-146.67	-152.52

APPENDIX D
SECTION PROPERTIES PROGRAM

APPENDIX D

Section Properties Program

Program Description

The section properties program calculates the area, centroid, radius to centroid, section moments and torsional constant of ring cross sections. Line segments parallel to the x- and y-axes define the cross section.

To compute the area, centroid, and section moments, the program divides the cross section into a coarse mesh. The section corner points determine the element edges. The area is then found by summing the elemental areas and the centroid is computed from the area weighted average of the x and y values of the element centroids. The moments for the straight beam theory are:

$$I_x = \int y^2 dA$$

$$I_y = \int x^2 dA \quad (1)$$

$$I_{xy} = \int_A xy dA$$

These are computed exactly by the program using the parallel axis theorem. The moments of inertia for curved beam theory are:

$$J_x = \int_A \frac{y^2}{1 - x/R} dA$$

$$J_y = \int_A \frac{x^2}{1 - x/R} dA \quad (2)$$

$$J_{xy} = \int_A \frac{xy}{1 - x/R} dA$$

These are also computed exactly.

To calculate the torsional constant, the program refines the mesh and then uses a finite difference scheme to solve the Poisson equation.

Program Input

The program requires that the cross section be oriented in an x-y coordinate system shown in the text. The x-coordinate is in the radial direction towards the center of curvature and the y-coordinate is in the axial direction. The following variables are input to the program:

XC--x-coordinates of the corner points. These must be input in clockwise order around the section.

YC--y-coordinates of the corner points. Their order must correspond to the x-coordinates.

RX1--Radius at first corner point

NC--Number of corner points

DXMAX--Maximum grid spacing desired in the x-direction for the refined mesh. The number of points along the x-direction must be 40 or less as the program is set up.

DYMAX--Maximum grid spacing desired in the y-direction

Sample Problem 1

The first sample problem is to find the section properties of the cross section shown in Figure D-1 using the program. The cross section is oriented in an x-y coordinate system as shown in Figure D-2. The corner point coordinates are

N	XC	YC
1	0.0	0.0
2	0.0	0.75
3	0.75	0.75
4	0.75	1.0
5	2.0	1.0
6	2.0	0.5
7	1.5	0.5
8	1.5	0.0
9	0.5	0.0
10	0.5	0.2
11	0.25	0.2
12	0.25	0.0

The radius at point 1 is 3.0 in and the chosen grid size is 0.2 in; so $R \times 1 = 3.0$, $NC = 12$, $DXMAX = 0.2$, and $DYMAX = 0.2$.

The results are shown in the program output. The section properties matrix and the torsional properties matrix are representations of the element meshes. Each "1" identifies a solid element of the section.

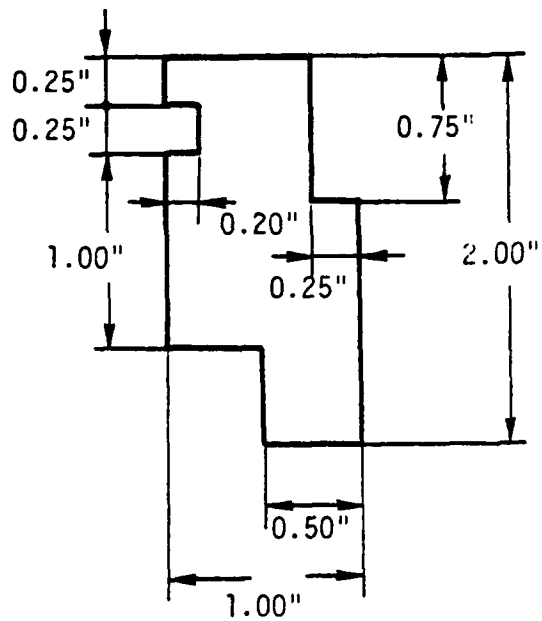


Figure D-1. Cross Section for First Sample Problem

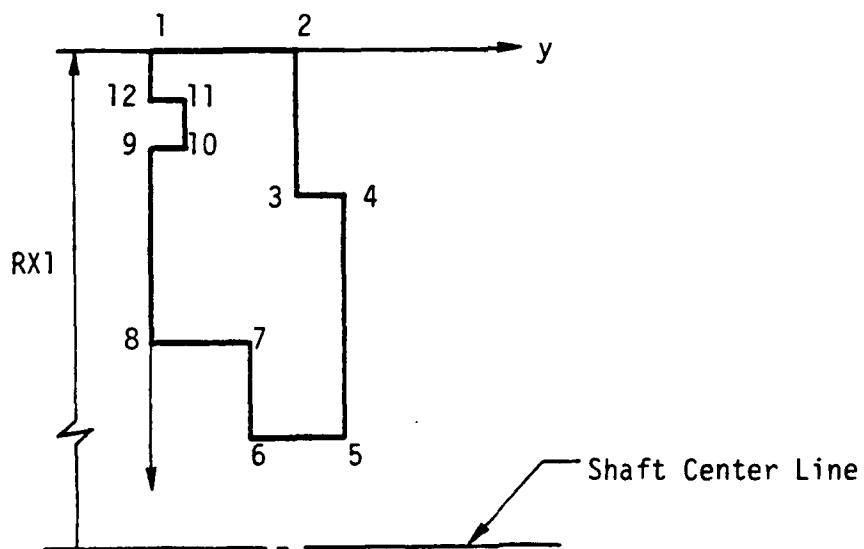


Figure D-2. Cross Section in Coordinate System.

Sample Problem 2

The second sample problem allows a comparison between the numerical and theoretical section property values. A rectangular section was used (Fig.D-3). The preparation of the program input is similar to the first problem.

The output is shown in the printout. The closed form solutions of equations (2) are obtained by integration. The section moments of inertia are:

$$J_x = -R \frac{(y_2^3 - y_1^3)}{3} \ln \left(\frac{R - x_2}{R - x_1} \right)$$

$$J_y = - (R^2)(y_2 - y_1) \left[\frac{1}{2} \left(\frac{x_2^2 - x_1^2}{R} \right) + (x_2 - x_1) + R \ln \left(\frac{R - x_2}{R - x_1} \right) \right]$$

$$J_{xy} = -R \left(\frac{y_2^2 - y_1^2}{2} \right) \left[(x_2 - x_1) + R \ln \left(\frac{R - x_2}{R - x_1} \right) \right]$$

Substituting in

$$x_1 = -0.25$$

$$x_2 = 0.25$$

$$y_1 = -0.5$$

$$y_2 = 0.5$$

$$R = 2.5$$

The moments of inertia are:

$$J_x = 0.04181 \text{ in}^4$$

$$J_y = 0.01048 \text{ in}^4$$

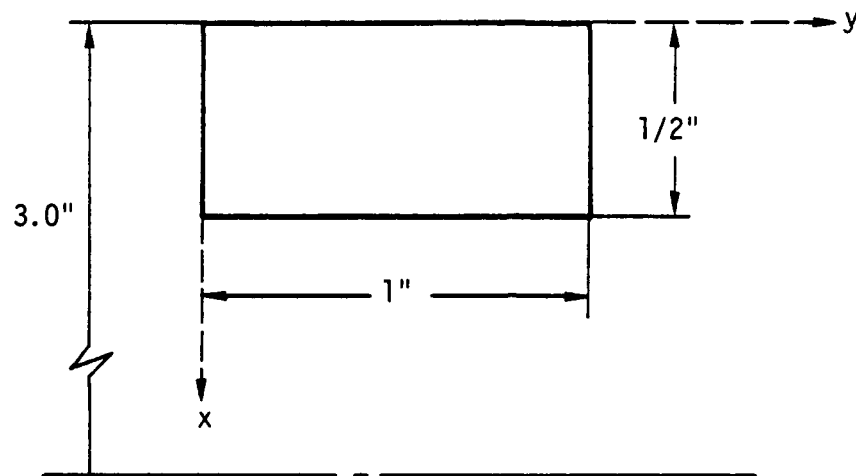


Figure D-3. Cross Section used for Comparing Numerical and Theoretical Section Moments.

$$J_{xy} = 0.0 \text{ in}^4$$

The expression for the torsion constant for a rectangular cross section is [38]

$$J = ab^3 \left[\frac{16}{3} - 3.36 \frac{b}{a} \left(1 - \frac{b^4}{12a^4} \right) \right]$$

where a and b are respectively half the length of the long and short sides of the rectangle. Substituting in

$$a = 0.5$$

$$b = 0.25$$

the torsion constant is

$$J = 0.02861 \text{ in}^4.$$

The numerical section moments of inertia are

$$J_x = 0.04178 \text{ in}^4$$

$$J_y = 0.01047 \text{ in}^4$$

$$J_{xy} = 0.0 \text{ in}^4$$

These agree to three significant figures with the theoretical results.

The numerical torsion constant is 0.02776 in^4 which differs by approximately 3% from the theoretical result.

SECTION 07/10/84 14.12 41

SAMPLE PROBLEM 2

INPUT GRID LOCATIONS

NO. 0 0 0 5000 0 5000 0 0 0 0 0
 0 0 0 0 0 0 0 0 0
 0 0 0 0
YC 0 1.0000 1.0000 0 0 0 0 0 0
 0 0 0 0 0 0 0 0 0
 0 0 0 0

ADJUSTED GRID LOCATIONS

NO. 0 0 5000
YC 0 1.0000

SECTION PROPERTIES MATRIX

0 0 0
0 1 0
0 0 0

AREA= 0.100E+01 PER= 0.1501E+00 COL= 0.000E+00 TH= 0.000E+00
I = 0.0107E+01 J = 0.1041E+01 K = 0.000E+00
L = 0.0173E+01 M = 0.1047E+01 N = 0.000E+00

FINER MESH ADJUSTMENT

NO. 0 0.0455 0.0909 0.1364 0.1818 0.2273 0.2727 0.3182 0.3636
 0.4091 0.4545 0.5000
YC 0 0.0909 0.1818 0.2727 0.3636 0.4545 0.5455 0.6364 0.7273
 0.8182 0.9091 1.0000

TORSIONAL PROPERTIES MATRIX

0 0 0 0 0 0 0 0 0 0 0 0 0 0
0 1 1 1 1 1 1 1 1 1 1 1 0
0 1 1 1 1 1 1 1 1 1 1 1 0
0 1 1 1 1 1 1 1 1 1 1 1 0
0 1 1 1 1 1 1 1 1 1 1 1 0
0 1 1 1 1 1 1 1 1 1 1 1 0
0 1 1 1 1 1 1 1 1 1 1 1 0
0 1 1 1 1 1 1 1 1 1 1 1 0
0 1 1 1 1 1 1 1 1 1 1 1 0
0 1 1 1 1 1 1 1 1 1 1 1 0
0 1 1 1 1 1 1 1 1 1 1 1 0
0 1 1 1 1 1 1 1 1 1 1 1 0
0 1 1 1 1 1 1 1 1 1 1 1 0
0 0 0 0 0 0 0 0 0 0 0 0 0

DIFFERENCE IN AREA= 0.2105E-05

THETA= 0.2775E-01

SECTION 07/20/84 14 14 47

SAMPLE PROBLEM 1

INPUT GRID LOCATIONS

XC.
0. 0. 0.7500 0.7500 2.0000 2.0000 1.5000 1.5000 0.5000
0.5000 0.2500 0.2500 0. 0. 0. 0. 0. 0.
0. 0. 0. 0.
YC.
0. 0.7500 0.7500 1.0000 1.0000 0.5000 0.5000 0. 0.
0.2000 0.2000 0. 0. 0. 0. 0. 0. 0.
0. 0. 0. 0.

ADJUSTED GRID LOCATIONS

XG
0. 0.2500 0.5000 0.7500 1.5000 2.0000
YG
0. 0.2000 0.5000 0.7500 1.0000

SECTION PROPERTIES MATRIX

0 0 0 0 0 0
0 1 1 1 0 0
0 0 1 1 0 0
0 1 1 1 0 0
0 1 1 1 1 0
0 0 0 1 1 0
0 0 0 0 0 0

AREA= 0.1311E+01 XBAR= 0.5742E+00 YBAR= 0.5081E+00 ZI= 1.1019E+11
IX= 0.1102E+00 IY= 0.4180E+00 IZ= 0.7188E+01
JX= 0.1128E+00 JY= 0.4846E+00 JZ= 0.1020E+00

FINER MESH ADJUSTMENT

XGF
0. 0.1250 0.2500 0.3750 0.5000 0.6250 0.7500 0.9000 1.0500
1.2000 1.3500 1.5000 1.6250 1.7500 1.8750 2.0000
YGF
0. 0.1000 0.2000 0.3000 0.4000 0.5000 0.6250 0.7500 0.8750
1.0000

TORSIONAL PROPERTIES MATRIX

0 0 0 0 0 0 0 0 0 0
0 1 1 1 1 1 1 1 0 0
0 1 1 1 1 1 1 1 0 0
0 0 0 1 1 1 1 1 0 0
0 0 0 1 1 1 1 1 0 0
0 1 1 1 1 1 1 1 0 0
0 1 1 1 1 1 1 1 0 0
0 1 1 1 1 1 1 1 1 0
0 1 1 1 1 1 1 1 1 0
0 1 1 1 1 1 1 1 1 0
0 1 1 1 1 1 1 1 1 0
0 1 1 1 1 1 1 1 1 0
0 0 0 0 0 0 1 1 1 0
0 0 0 0 0 0 1 1 1 0
0 0 0 0 0 0 1 1 1 0
0 0 0 0 0 0 1 1 1 0
0 0 0 0 0 0 0 0 0 0

DIFFERENCE IN AREA= 0.1144E-04

JTHETA= 0.1754E+00

STOP
TIME 0.5 SECS


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10 C SECTION PROPERTIES PROGRAM
20 C THIS PROGRAM CALCULATES THE SECTION PROPERTIES OF
30 C CROSS SECTIONS MADE UP OF ANY NUMBER OF LINE SEG-
40 C MENTS PARALLEL TO THE X AND Y AXES. (THE X-AXIS IS
50 C IN THE RADIAL DIRECTION TOWARDS THE CENTER OF CURVA-
60 C TURE AND THE Y-AXIS IS IN THE AXIAL DIRECTION.
70 C THIS PROGRAM COMPUTES THE CROSS SECTIONAL AREA, THE
80 C SECTION CENTROID AND THE RADIUS AT THE CENTROID. IT
90 C COMPUTES THE SECTION MOMENTS USING BOTH STRAIGHT BEAM
100 C AND CURVED BEAM THEORY. IT ALSO CALCULATES THE TOR-
110 C SIONAL CONSTANT.
120 C
130 C DOUBLE PRECISION XG(22),YG(22),XGP(40),YGP(40),XC(22),YC(22)
140 C DOUBLE PRECISION AIM1(40,40),AIF1(40,40),AJM1(40,40),AJF1(40,40)
150 C DOUBLE PRECISION P(40,40),EPHS(40,40),FOLD,VAL
160 C DIMENSION IEO(22),IGXC(22),JG+C(22),ISOLID(22,22),IGN(40)
170 C DIMENSION JGN(40),IPSO(1,40,40)
180 C DIMENSION ZFE(80),YFE(80)
190 C CALL OPSYS('ALLOC', 'FEData', 7)
200 C
210 C INPUT THE COORDINATES OF THE CORNER POINTS IN CLOCKWISE
220 C ORDER. THE STARTING POINT IS ARBITRARY.
230 C DATA XG(0,0,0,0,0,75,0,75,2,0,2,0,1,3,1,5,1,5,5,25,25,7
240 C DATA YG(0,0,0,0,75,0,75,1,0,1,0,0,3,0,5,0,0,0,0,2,0,2,0,0,
250 C
260 C INPUT RADIUS AT X1 AND NUMBER OF CORNER POINTS
270 C RX1=0.0
280 C NC=12
290 C NCH=1
300 C
310 C INPUT GRID SIZE FOR FINER MESH
320 C IMA=0.1
330 C IMA=0.1
340 C
350 C SET UP THE GRID POINTS
360 C NG=NC/2
370 C DO 10 I=1,NG
380 C   NG(I)=NC(2*I)
390 C   10   YG(I)=YG(2*I)
400 C
410 C ARRANGE THE GRID POINTS IN INCREASING ORDER
420 C DO 20 I=1,NG
430 C   IET=1
440 C   AMIN=1.E+06
450 C   DO 21 I=1,NG
460 C     IF YG(I) < AMIN GO TO 22
470 C     AMIN=YG(I)
480 C     I=I
490 C   21 CONTINUE
500 C   YTRF=YG(I)
510 C   DO 23 I=NG,1
520 C     IF I=IET GO TO 25
530 C     I=I-1
540 C   23 CONTINUE
550 C   YG(I)=YTRF
560 C   25 YG(I)=YTRF
570 C
580 C
590 C
600 C
610 C
620 C
630 C
640 C

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650 C ELIMINATE EQUAL NG AND NG VALUES
660     NGM1=NG-1
670     IIEQ=0
680     DO 10 I=1,NGM1
690         IF(XG(I).NE.XG(I+1)) GO TO 10
700         IIEQ=IIEQ+1
710         IEQ(IIEQ)=I
720     DO CONTINUE
730     NGX=NG
740     IF(IIEQ.EQ.0) GO TO 100
750     DO 40 I=1,IIEQ
760         NGX=NGX-1
770         IST=IEQ(I)
780         DO 50 II=IST,NGX
790             NG(II)=XG(II+1)
800     40 CONTINUE
810     100 IIEQ=0
820     DO 60 I=1,NGM1
830         IF(YG(I).NE.YG(I+1)) GO TO 60
840         IIEQ=IIEQ+1
850         IEQ(IIEQ)=I
860     DO CONTINUE
870     NGY=NG
880     IF(IIEQ.EQ.0) GO TO 200
890     DO 70 I=1,IIEQ
900         NGY=NGY-1
910         IST=IEQ(I)
920         DO 80 II=IST,NGY
930             YG(II)=YG(II+1)
940     70 CONTINUE
950 C
960 C MATCH CORNER POINTS WITH GRID POINTS
970 C IGVC(10) IS GRID POINT CORRESPONDING TO CORNER POINT 10
980     DO 90 IC=1,NC
990         DO 100 IC=1,NC
1000             IF(XC(10).EQ.XG(10) .AND. YC(10).EQ.YG(10))
1010                 GO CONTINUE
1020             DO 95 JC=1,NG
1030                 DO 95 JG=1,NGY
1040                     IF(YC(JC).EQ.YG(JG)) JGYC(JC)=JG
1050             95 CONTINUE
1060 C
1070 C IDENTIFY THE VOID AND SOLID REGIONS
1080 C I,J REFERS TO REGION WHICH IS ABOVE AND TO THE LEFT OF POINT I,J
1090 C CODE: VOID=0, SOLID=1, UNASSIGNED=2
1100     NGMP1=NGM+1
1110     NGYP1=NGY+1
1120     DO 101 I=1,NGMP1
1130         DO 101 J=1,NGYP1
1140             101 ISOLID(I,J)=2
1150     DO 1000 M=1,NC
1160         M=M+1
1170         IF(M.EQ.NC) M=1
1180         IC=0
1190         IF(XC(M).GT.XC(M+1)) IC=1
1200         IF(XC(M).GT.XC(M)) IC=2
1210         IF(XC(M).LT.XC(M+1)) IC=3
1220         IF(XC(M).LT.XC(M)) IC=4
1230         GO TO (110,120,130,140),IC
1240     110     IN=IGNC(N)
1250             JN=JGYC(N)
1260             JM=JGYC(M)
1270             JM=JM+1
1280             DO 112 J=JM,JN
1290                 ISOLID(IN,J)=0
1300     112     ISOLID(IN+1,J)=1

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1310      GO TO 1000
1320      IM=IGXC(N)
1330      IM=IM+1
1340      JM=JGYC(N)
1350      IN=IGXC(N)
1360      DO 122 I=IM, IN
1370          ISOLID(I, JM)=1
1380      122      ISOLID(I, JM+1)=0
1390      GO TO 1000
1400      IN=IGXC(N)
1410      JN=JGYC(N)
1420      JN=JN+1
1430      JM=JGYC(N)
1440      DO 132 J=JN, JM
1450          ISOLID(IN, J)=1
1460      132      ISOLID(IN+1, J)=0
1470      GO TO 1000
1480      IN=IGXC(N)
1490      IN=IN+1
1500      JN=JGYC(N)
1510      IN=IGXC(N)
1520      DO 142 I=IN, IN
1530          ISOLID(I, JN)=0
1540      142      ISOLID(I, JN+1)=1
1550      1000 CONTINUE
1560 C
1570 C SET BOUNDARIES TO VOID
1580      DO 310 J=1, NGVP1
1590          ISOLID(1, J)=0
1600      310      ISOLID(NGXP1, J)=0
1610      DO 320 I=1, NGXP1
1620          ISOLID(I, 1)=0
1630      320      ISOLID(I, NGVP1)=0
1640 C
1650 C USE X SWEEP TO SET ALL OTHERS
1660      DO 400 I=1, NGX
1670          DO 410 J=1, NGVP1
1680              IF (ISOLID(I, J).NE.2) GO TO 410
1690              J1=J
1700              DO 420 JJ=J1, NGVP1
1710                  IF (ISOLID(I, JJ).EQ.2) GO TO 420
1720                  J2=JJ-1
1730                  GO TO 430
1740      420      CONTINUE
1750              WRITE(6,11)
1760      11      FORMAT(1H 'TROUBLE1')
1770      400      DO 405 JJ=J1, J2
1780          405      ISOLID(I, JJ)=ISOLID(I, J1-1)
1790      410      CONTINUE
1800      400 CONTINUE
1810 C
1820 C DETERMINE CENTROID LOCATION
1830      SM=0.
1840      SMY=0.
1850      AREA=0.
1860      DO 500 I=1, NGX
1870          DO 500 J=1, NGVP
1880              ISOL=ISOLID(I, J)
1890              DA=(XG(I)-XG(I-1))*(YG(J)-YG(J-1))*ISOL
1900              AREA=AREA+DA
1910              X=(XG(I)-XG(I-1))* 5
1920              Y=(YG(J)-YG(J-1))* 5
1930              SMY=SMY+DA*X
1940      500      SMX=SMX+DA*Y
1950      XBAR=SMY/AREA
1960      YBAR=SMX/AREA

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1970 C
1980 C DETERMINE SECTION MOMENTS
1990     AIX=0.0
2000     AIY=0.0
2010     AIXY=0.0
2020     AJX=0.0
2030     AJY=0.0
2040     AJXY=0.0
2050     RC=RX1+XG(I)-XEAR
2060     DO 510 I=2,NGX
2070         DO 510 J=2,NGY
2080             ISOL=ISOLID(I,J)
2090             DX=XG(I)-XG(I-1)
2100             DY=YG(J)-YG(J-1)
2110             DA=DX*DY
2120             X=(XG(I)+XG(I-1))*0.5
2130             Y=(YG(J)+YG(J-1))*0.5
2140             X=X-XEAR
2150             Y=Y-YEAR
2160             X1=X-DX*0.5
2170             X2=X+DX*0.5
2180             Y1=Y-DY*0.5
2190             Y2=Y+DY*0.5
2200             AIX=AIX+(DX*DY*(3/12 +DA*(Y+Y1))*ISOL
2210             AIY=AIY+(DY*DX*(3/12 +DA*(X+X1))*ISOL
2220             AIXY=AIXY+X*Y*DA*ISOL
2230             FACT=ALOG((RC-X2)/(RC-X1))
2240             AJX=AJX-RC*(Y2+Y-Y1**3)/12 +0*FACT*ISOL
2250             TERM1=0.5*(X2+X-X1**2)/RC
2260             TERM2=X2-X1+RC*ALOG((RC-X2)/(RC-X1))
2270             AJY=AJY-RC*(X2-X1)*(Y2-Y1)*(TERM1+TERM2)*ISOL
2280             TERM1=X2-X1+RC*ALOG((RC-X2)/(RC-X1))
2290             AJXY=AJXY-RC*(Y2+Y-Y1**2)/2*TERM1*ISOL
2300     510 CONTINUE
2310 C
2320 C REFINES THE MESH
2330     JY=0
2340     IX=0
2350     NGXM1=NGX-1
2360     DO 540 I=1,NGXM1
2370         IGN(I)=(XG(I+1)-XG(I))/DXMAX+1.5
2380         DXUSED=(XG(I+1)-XG(I))/IGN(I)
2390         IGNI=IGN(I)
2400         IF(I.EQ. NGXM1) IGNI=IGNI+1
2410         DO 520 II=1,IGNI
2420             IX=IX+1
2430             XGP(IX)=XG(I)+DXUSED*(II-1)
2440     520 CONTINUE
2450     540 CONTINUE
2460     NGYM1=NGY-1
2470     DO 550 I=1,NGYM1
2480         JGN(I)=(YG(I+1)-YG(I))/DYMAX+1.5
2490         DYUSED=(YG(I+1)-YG(I))/JGN(I)
2500         JGNI=JGN(I)
2510         IF(I.EQ. NGYM1) JGNI=JGNI+1
2520         DO 530 JJ=1,JGNI
2530             JY=JY+1
2540             YGP(JY)=YG(I)+DYUSED*(JJ-1)
2550     530 CONTINUE
2560     550 CONTINUE
2570 C
2580 C SET ADDITIONAL MESH AREAS TO SOLID OR VOID.
2590     IXMAN=IX
2600     JYMAN=JY
2610     JSEG=1
2620     ISEG=1

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2630      NFGNPF1=INMPN1-1
2640      NF3YF1=JYMAX+1
2650      DO 560 I1=1,NFGNPF1
2660          DO 580 JJ=1,NF3YF1
2670              560      IFSOLI(I1,JJ)=0.
2680      DO 595 I=2,NGN
2690          JBEG=1
2700          DO 580 J=2,NGY
2710              IEND=JBEG+IGN(I-1)
2720              JEND=JBEG+JGN(J-1)
2730              IBEGF1=JBEG+1
2740              JBEGF1=JBEG+1
2750              IF (ISOLID(I,J).EQ.1) GO TO 570
2760              GO TO 590
2770      570      CONTINUE
2780          DO 580 I1=IBEGF1,IEND
2790              DO 580 JJ=JBEGF1,JEND
2800                  580      IFSOLI(I1,JJ)=1
2810      590      JBEG=JGN(J-1)+JBEG
2820      595      IBEG=IGN(I-1)+IBEG
2830 C
2840 C CHECK TO SEE IF TOTAL AREA IS THE SAME.
2850      AREA2=0
2860      DO 700 I=1,IMAX
2870          DO 700 J=2,JYMAX
2880              IFSOL=IFSOLI(I,J)
2890              DA=(XGF(I)-XGF(I-1))*(YGF(J)-YGF(J-1))+IFSOL
2900              AREA2=AREA2+DA
2910      700 CONTINUE
2920      DIFF=AREA-AREA2
2930 C
2940 C PRINT OUT RESULTS
2950      WRITE (6,18)
2960      18 FORMAT('***** SAMPLE DATA *****')
2970      WRITE (6,19)
2980      19 FORMAT('***** INPUT GRID LOCATIONS *****')
2990      WRITE (6,20) NO
3000      20 FORMAT('NO',F7.4)
3010      WRITE (6,21)
3020      21 FORMAT('NO',F7.4)
3030      WRITE (6,22) NO
3040      22 FORMAT('NO',F7.4)
3050      WRITE (6,23) NO
3060      23 FORMAT('NO',F7.4)
3070      WRITE (6,24) NO
3080      24 FORMAT('NO',F7.4)
3090      WRITE (6,25) NO
3100      25 FORMAT('NO',F7.4)
3110      WRITE (6,26) NO
3120      26 FORMAT('NO',F7.4)
3130      WRITE (6,27) NO
3140      27 FORMAT('NO',F7.4)
3150      WRITE (6,28) NO
3160      28 FORMAT('NO',F7.4)
3170      WRITE (6,29) NO
3180      29 FORMAT('NO',F7.4)
3190      WRITE (6,30) NO
3200      30 FORMAT('NO',F7.4)
3210      WRITE (6,31) NO
3220      31 FORMAT('NO',F7.4)
3230      WRITE (6,32) NO
3240      32 FORMAT('NO',F7.4)
3250      WRITE (6,33) NO
3260      33 FORMAT('NO',F7.4)
3270      WRITE (6,34) NO
3280      34 FORMAT('NO',F7.4)
3290      WRITE (6,35) NO
3300      35 FORMAT('NO',F7.4)
3310      WRITE (6,36) NO
3320      36 FORMAT('NO',F7.4)
3330      WRITE (6,37) NO
3340      37 FORMAT('NO',F7.4)
3350      WRITE (6,38) NO
3360      38 FORMAT('NO',F7.4)
3370      WRITE (6,39) NO
3380      39 FORMAT('NO',F7.4)
3390      WRITE (6,40) NO
3400      40 FORMAT('NO',F7.4)
3410      WRITE (6,41) NO
3420      41 FORMAT('NO',F7.4)
3430      WRITE (6,42) NO
3440      42 FORMAT('NO',F7.4)
3450      WRITE (6,43) NO
3460      43 FORMAT('NO',F7.4)
3470      WRITE (6,44) NO
3480      44 FORMAT('NO',F7.4)
3490      WRITE (6,45) NO
3500      45 FORMAT('NO',F7.4)
3510      WRITE (6,46) NO
3520      46 FORMAT('NO',F7.4)
3530      WRITE (6,47) NO
3540      47 FORMAT('NO',F7.4)
3550      WRITE (6,48) NO
3560      48 FORMAT('NO',F7.4)
3570      WRITE (6,49) NO
3580      49 FORMAT('NO',F7.4)
3590      WRITE (6,50) NO
3600      50 FORMAT('NO',F7.4)
3610      WRITE (6,51) NO
3620      51 FORMAT('NO',F7.4)
3630      WRITE (6,52) NO
3640      52 FORMAT('NO',F7.4)
3650      WRITE (6,53) NO
3660      53 FORMAT('NO',F7.4)
3670      WRITE (6,54) NO
3680      54 FORMAT('NO',F7.4)
3690      WRITE (6,55) NO
3700      55 FORMAT('NO',F7.4)
3710      WRITE (6,56) NO
3720      56 FORMAT('NO',F7.4)
3730      WRITE (6,57) NO
3740      57 FORMAT('NO',F7.4)
3750      WRITE (6,58) NO
3760      58 FORMAT('NO',F7.4)
3770      WRITE (6,59) NO
3780      59 FORMAT('NO',F7.4)
3790      WRITE (6,60) NO
3800      60 FORMAT('NO',F7.4)
3810      WRITE (6,61) NO
3820      61 FORMAT('NO',F7.4)
3830      WRITE (6,62) NO
3840      62 FORMAT('NO',F7.4)
3850      WRITE (6,63) NO
3860      63 FORMAT('NO',F7.4)
3870      WRITE (6,64) NO
3880      64 FORMAT('NO',F7.4)
3890      WRITE (6,65) NO
3900      65 FORMAT('NO',F7.4)
3910      WRITE (6,66) NO
3920      66 FORMAT('NO',F7.4)
3930      WRITE (6,67) NO
3940      67 FORMAT('NO',F7.4)
3950      WRITE (6,68) NO
3960      68 FORMAT('NO',F7.4)
3970      WRITE (6,69) NO
3980      69 FORMAT('NO',F7.4)
3990      WRITE (6,70) NO
4000      70 FORMAT('NO',F7.4)
4010      WRITE (6,71) NO
4020      71 FORMAT('NO',F7.4)
4030      WRITE (6,72) NO
4040      72 FORMAT('NO',F7.4)
4050      WRITE (6,73) NO
4060      73 FORMAT('NO',F7.4)
4070      WRITE (6,74) NO
4080      74 FORMAT('NO',F7.4)
4090      WRITE (6,75) NO
4100      75 FORMAT('NO',F7.4)
4110      WRITE (6,76) NO
4120      76 FORMAT('NO',F7.4)
4130      WRITE (6,77) NO
4140      77 FORMAT('NO',F7.4)
4150      WRITE (6,78) NO
4160      78 FORMAT('NO',F7.4)
4170      WRITE (6,79) NO
4180      79 FORMAT('NO',F7.4)
4190      WRITE (6,80) NO
4200      80 FORMAT('NO',F7.4)
4210      WRITE (6,81) NO
4220      81 FORMAT('NO',F7.4)
4230      WRITE (6,82) NO
4240      82 FORMAT('NO',F7.4)
4250      WRITE (6,83) NO
4260      83 FORMAT('NO',F7.4)
4270      WRITE (6,84) NO
4280      84 FORMAT('NO',F7.4)
4290      WRITE (6,85) NO
4300      85 FORMAT('NO',F7.4)
4310      WRITE (6,86) NO
4320      86 FORMAT('NO',F7.4)
4330      WRITE (6,87) NO
4340      87 FORMAT('NO',F7.4)
4350      WRITE (6,88) NO
4360      88 FORMAT('NO',F7.4)
4370      WRITE (6,89) NO
4380      89 FORMAT('NO',F7.4)
4390      WRITE (6,90) NO
4400      90 FORMAT('NO',F7.4)
4410      WRITE (6,91) NO
4420      91 FORMAT('NO',F7.4)
4430      WRITE (6,92) NO
4440      92 FORMAT('NO',F7.4)
4450      WRITE (6,93) NO
4460      93 FORMAT('NO',F7.4)
4470      WRITE (6,94) NO
4480      94 FORMAT('NO',F7.4)
4490      WRITE (6,95) NO
4500      95 FORMAT('NO',F7.4)
4510      WRITE (6,96) NO
4520      96 FORMAT('NO',F7.4)
4530      WRITE (6,97) NO
4540      97 FORMAT('NO',F7.4)
4550      WRITE (6,98) NO
4560      98 FORMAT('NO',F7.4)
4570      WRITE (6,99) NO
4580      99 FORMAT('NO',F7.4)
4590      WRITE (6,100) NO
4600      100 FORMAT('NO',F7.4)
4610      WRITE (6,101) NO
4620      101 FORMAT('NO',F7.4)
4630      WRITE (6,102) NO
4640      102 FORMAT('NO',F7.4)
4650      WRITE (6,103) NO
4660      103 FORMAT('NO',F7.4)
4670      WRITE (6,104) NO
4680      104 FORMAT('NO',F7.4)
4690      WRITE (6,105) NO
4700      105 FORMAT('NO',F7.4)
4710      WRITE (6,106) NO
4720      106 FORMAT('NO',F7.4)
4730      WRITE (6,107) NO
4740      107 FORMAT('NO',F7.4)
4750      WRITE (6,108) NO
4760      108 FORMAT('NO',F7.4)
4770      WRITE (6,109) NO
4780      109 FORMAT('NO',F7.4)
4790      WRITE (6,110) NO
4800      110 FORMAT('NO',F7.4)
4810      WRITE (6,111) NO
4820      111 FORMAT('NO',F7.4)
4830      WRITE (6,112) NO
4840      112 FORMAT('NO',F7.4)
4850      WRITE (6,113) NO
4860      113 FORMAT('NO',F7.4)
4870      WRITE (6,114) NO
4880      114 FORMAT('NO',F7.4)
4890      WRITE (6,115) NO
4900      115 FORMAT('NO',F7.4)
4910      WRITE (6,116) NO
4920      116 FORMAT('NO',F7.4)
4930      WRITE (6,117) NO
4940      117 FORMAT('NO',F7.4)
4950      WRITE (6,118) NO
4960      118 FORMAT('NO',F7.4)
4970      WRITE (6,119) NO
4980      119 FORMAT('NO',F7.4)
4990      WRITE (6,120) NO
5000      120 FORMAT('NO',F7.4)
5010      WRITE (6,121) NO
5020      121 FORMAT('NO',F7.4)
5030      WRITE (6,122) NO
5040      122 FORMAT('NO',F7.4)
5050      WRITE (6,123) NO
5060      123 FORMAT('NO',F7.4)
5070      WRITE (6,124) NO
5080      124 FORMAT('NO',F7.4)
5090      WRITE (6,125) NO
5100      125 FORMAT('NO',F7.4)
5110      WRITE (6,126) NO
5120      126 FORMAT('NO',F7.4)
5130      WRITE (6,127) NO
5140      127 FORMAT('NO',F7.4)
5150      WRITE (6,128) NO
5160      128 FORMAT('NO',F7.4)
5170      WRITE (6,129) NO
5180      129 FORMAT('NO',F7.4)
5190      WRITE (6,130) NO
5200      130 FORMAT('NO',F7.4)
5210      WRITE (6,131) NO
5220      131 FORMAT('NO',F7.4)
5230      WRITE (6,132) NO
5240      132 FORMAT('NO',F7.4)
5250      WRITE (6,133) NO
5260      133 FORMAT('NO',F7.4)
5270      WRITE (6,134) NO
5280      134 FORMAT('NO',F7.4)
5290      WRITE (6,135) NO
5300      135 FORMAT('NO',F7.4)
5310      WRITE (6,136) NO
5320      136 FORMAT('NO',F7.4)
5330      WRITE (6,137) NO
5340      137 FORMAT('NO',F7.4)
5350      WRITE (6,138) NO
5360      138 FORMAT('NO',F7.4)
5370      WRITE (6,139) NO
5380      139 FORMAT('NO',F7.4)
5390      WRITE (6,140) NO
5400      140 FORMAT('NO',F7.4)
5410      WRITE (6,141) NO
5420      141 FORMAT('NO',F7.4)
5430      WRITE (6,142) NO
5440      142 FORMAT('NO',F7.4)
5450      WRITE (6,143) NO
5460      143 FORMAT('NO',F7.4)
5470      WRITE (6,144) NO
5480      144 FORMAT('NO',F7.4)
5490      WRITE (6,145) NO
5500      145 FORMAT('NO',F7.4)
5510      WRITE (6,146) NO
5520      146 FORMAT('NO',F7.4)
5530      WRITE (6,147) NO
5540      147 FORMAT('NO',F7.4)
5550      WRITE (6,148) NO
5560      148 FORMAT('NO',F7.4)
5570      WRITE (6,149) NO
5580      149 FORMAT('NO',F7.4)
5590      WRITE (6,150) NO
5600      150 FORMAT('NO',F7.4)
5610      WRITE (6,151) NO
5620      151 FORMAT('NO',F7.4)
5630      WRITE (6,152) NO
5640      152 FORMAT('NO',F7.4)
5650      WRITE (6,153) NO
5660      153 FORMAT('NO',F7.4)
5670      WRITE (6,154) NO
5680      154 FORMAT('NO',F7.4)
5690      WRITE (6,155) NO
5700      155 FORMAT('NO',F7.4)
5710      WRITE (6,156) NO
5720      156 FORMAT('NO',F7.4)
5730      WRITE (6,157) NO
5740      157 FORMAT('NO',F7.4)
5750      WRITE (6,158) NO
5760      158 FORMAT('NO',F7.4)
5770      WRITE (6,159) NO
5780      159 FORMAT('NO',F7.4)
5790      WRITE (6,160) NO
5800      160 FORMAT('NO',F7.4)
5810      WRITE (6,161) NO
5820      161 FORMAT('NO',F7.4)
5830      WRITE (6,162) NO
5840      162 FORMAT('NO',F7.4)
5850      WRITE (6,163) NO
5860      163 FORMAT('NO',F7.4)
5870      WRITE (6,164) NO
5880      164 FORMAT('NO',F7.4)
5890      WRITE (6,165) NO
5900      165 FORMAT('NO',F7.4)
5910      WRITE (6,166) NO
5920      166 FORMAT('NO',F7.4)
5930      WRITE (6,167) NO
5940      167 FORMAT('NO',F7.4)
5950      WRITE (6,168) NO
5960      168 FORMAT('NO',F7.4)
5970      WRITE (6,169) NO
5980      169 FORMAT('NO',F7.4)
5990      WRITE (6,170) NO
6000      170 FORMAT('NO',F7.4)
6010      WRITE (6,171) NO
6020      171 FORMAT('NO',F7.4)
6030      WRITE (6,172) NO
6040      172 FORMAT('NO',F7.4)
6050      WRITE (6,173) NO
6060      173 FORMAT('NO',F7.4)
6070      WRITE (6,174) NO
6080      174 FORMAT('NO',F7.4)
6090      WRITE (6,175) NO
6100      175 FORMAT('NO',F7.4)
6110      WRITE (6,176) NO
6120      176 FORMAT('NO',F7.4)
6130      WRITE (6,177) NO
6140      177 FORMAT('NO',F7.4)
6150      WRITE (6,178) NO
6160      178 FORMAT('NO',F7.4)
6170      WRITE (6,179) NO
6180      179 FORMAT('NO',F7.4)
6190      WRITE (6,180) NO
6200      180 FORMAT('NO',F7.4)
6210      WRITE (6,181) NO
6220      181 FORMAT('NO',F7.4)
6230      WRITE (6,182) NO
6240      182 FORMAT('NO',F7.4)
6250      WRITE (6,183) NO
6260      183 FORMAT('NO',F7.4)
6270      WRITE (6,184) NO
6280      184 FORMAT('NO',F7.4)
6290      WRITE (6,185) NO
6300      185 FORMAT('NO',F7.4)
6310      WRITE (6,186) NO
6320      186 FORMAT('NO',F7.4)
6330      WRITE (6,187) NO
6340      187 FORMAT('NO',F7.4)
6350      WRITE (6,188) NO
6360      188 FORMAT('NO',F7.4)
6370      WRITE (6,189) NO
6380      189 FORMAT('NO',F7.4)
6390      WRITE (6,190) NO
6400      190 FORMAT('NO',F7.4)
6410      WRITE (6,191) NO
6420      191 FORMAT('NO',F7.4)
6430      WRITE (6,192) NO
6440      192 FORMAT('NO',F7.4)
6450      WRITE (6,193) NO
6460      193 FORMAT('NO',F7.4)
6470      WRITE (6,194) NO
6480      194 FORMAT('NO',F7.4)
6490      WRITE (6,195) NO
6500      195 FORMAT('NO',F7.4)
6510      WRITE (6,196) NO
6520      196 FORMAT('NO',F7.4)
6530      WRITE (6,197) NO
6540      197 FORMAT('NO',F7.4)
6550      WRITE (6,198) NO
6560      198 FORMAT('NO',F7.4)
6570      WRITE (6,199) NO
6580      199 FORMAT('NO',F7.4)
6590      WRITE (6,200) NO
6600      200 FORMAT('NO',F7.4)
6610      WRITE (6,201) NO
6620      201 FORMAT('NO',F7.4)
6630      WRITE (6,202) NO
6640      202 FORMAT('NO',F7.4)
6650      WRITE (6,203) NO
6660      203 FORMAT('NO',F7.4)
6670      WRITE (6,204) NO
6680      204 FORMAT('NO',F7.4)
6690      WRITE (6,205) NO
6700      205 FORMAT('NO',F7.4)
6710      WRITE (6,206) NO
6720      206 FORMAT('NO',F7.4)
6730      WRITE (6,207) NO
6740      207 FORMAT('NO',F7.4)
6750      WRITE (6,208) NO
6760      208 FORMAT('NO',F7.4)
6770      WRITE (6,209) NO
6780      209 FORMAT('NO',F7.4)
6790      WRITE (6,210) NO
6800      210 FORMAT('NO',F7.4)
6810      WRITE (6,211) NO
6820      211 FORMAT('NO',F7.4)
6830      WRITE (6,212) NO
6840      212 FORMAT('NO',F7.4)
6850      WRITE (6,213) NO
6860      213 FORMAT('NO',F7.4)
6870      WRITE (6,214) NO
6880      214 FORMAT('NO',F7.4)
6890      WRITE (6,215) NO
6900      215 FORMAT('NO',F7.4)
6910      WRITE (6,216) NO
6920      216 FORMAT('NO',F7.4)
6930      WRITE (6,217) NO
6940      217 FORMAT('NO',F7.4)
6950      WRITE (6,218) NO
6960      218 FORMAT('NO',F7.4)
6970      WRITE (6,219) NO
6980      219 FORMAT('NO',F7.4)
6990      WRITE (6,220) NO
7000      220 FORMAT('NO',F7.4)
7010      WRITE (6,221) NO
7020      221 FORMAT('NO',F7.4)
7030      WRITE (6,222) NO
7040      222 FORMAT('NO',F7.4)
7050      WRITE (6,223) NO
7060      223 FORMAT('NO',F7.4)
7070      WRITE (6,224) NO
7080      224 FORMAT('NO',F7.4)
7090      WRITE (6,225) NO
7100      225 FORMAT('NO',F7.4)
7110      WRITE (6,226) NO
7120      226 FORMAT('NO',F7.4)
7130      WRITE (6,227) NO
7140      227 FORMAT('NO',F7.4)
7150      WRITE (6,228) NO
7160      228 FORMAT('NO',F7.4)
7170      WRITE (6,229) NO
7180      229 FORMAT('NO',F7.4)
7190      WRITE (6,230) NO
7200      230 FORMAT('NO',F7.4)
7210      WRITE (6,231) NO
7220      231 FORMAT('NO',F7.4)
7230      WRITE (6,232) NO
7240      232 FORMAT('NO',F7.4)
7250      WRITE (6,233) NO
7260      233 FORMAT('NO',F7.4)
7270      WRITE (6,234) NO
7280      234 FORMAT('NO',F7.4)
7290      WRITE (6,235) NO
7300      235 FORMAT('NO',F7.4)
7310      WRITE (6,236) NO
7320      236 FORMAT('NO',F7.4)
7330      WRITE (6,237) NO
7340      237 FORMAT('NO',F7.4)
7350      WRITE (6,238) NO
7360      238 FORMAT('NO',F7.4)
7370      WRITE (6,239) NO
7380      239 FORMAT('NO',F7.4)
7390      WRITE (6,240) NO
7400      240 FORMAT('NO',F7.4)
7410      WRITE (6,241) NO
7420      241 FORMAT('NO',F7.4)
7430      WRITE (6,242) NO
7440      242 FORMAT('NO',F7.4)
7450      WRITE (6,243) NO
7460      243 FORMAT('NO',F7.4)
7470      WRITE (6,244) NO
7480      244 FORMAT('NO',F7.4)
7490      WRITE (6,245) NO
7500      245 FORMAT('NO',F7.4)
7510      WRITE (6,246) NO
7520      246 FORMAT('NO',F7.4)
7530      WRITE (6,247) NO
7540      247 FORMAT('NO',F7.4)
7550      WRITE (6,248) NO
7560      248 FORMAT('NO',F7.4)
7570      WRITE (6,249) NO
7580      249 FORMAT('NO',F7.4)
7590      WRITE (6,250) NO
7600      250 FORMAT('NO',F7.4)
7610      WRITE (6,251) NO
7620      251 FORMAT('NO',F7.4)
7630      WRITE (6,252) NO
7640      252 FORMAT('NO',F7.4)
7650      WRITE (6,253) NO
7660      253 FORMAT('NO',F7.4)
7670      WRITE (6,254) NO
7680      254 FORMAT('NO',F7.4)
7690      WRITE (6,255) NO
7700      255 FORMAT('NO',F7.4)
7710      WRITE (6,256) NO
7720      256 FORMAT('NO',F7.4)
7730      WRITE (6,257) NO
7740      257 FORMAT('NO',F7.4)
7750      WRITE (6,258) NO
7760      258 FORMAT('NO',F7.4)
7770      WRITE (6,259) NO
7780      259 FORMAT('NO',F7.4)
7790      WRITE (6,260) NO
7800      260 FORMAT('NO',F7.4)
7810      WRITE (6,261) NO
7820      261 FORMAT('NO',F7.4)
7830      WRITE (6,262) NO
7840      262 FORMAT('NO',F7.4)
7850      WRITE (6,263) NO
7860      263 FORMAT('NO',F7.4)
7870      WRITE (6,264) NO
7880      264 FORMAT('NO',F7.4)
7890      WRITE (6,265) NO
7900      265 FORMAT('NO',F7.4)
7910      WRITE (6,266) NO
7920      266 FORMAT('NO',F7.4)
7930      WRITE (6,267) NO
7940      267 FORMAT('NO',F7.4)
7950      WRITE (6,268) NO
7960      268 FORMAT('NO',F7.4)
7970      WRITE (6,269) NO
7980      269 FORMAT('NO',F7.4)
7990      WRITE (6,270) NO
8000      270 FORMAT('NO',F7.4)
8010      WRITE (6,271) NO
8020      271 FORMAT('NO',F7.4)
8030      WRITE (6,272) NO
8040      272 FORMAT('NO',F7.4)
8050      WRITE (6,273) NO
8060      273 FORMAT('NO',F7.4)
8070      WRITE (6,274) NO
8080      274 FORMAT('NO',F7.4)
8090      WRITE (6,275) NO
8100      275 FORMAT('NO',F7.4)
8110      WRITE (6,276) NO
8120      276 FORMAT('NO',F7.4)
8130      WRITE (6,277) NO
8140      277 FORMAT('NO',F7.4)
8150      WRITE (6,278) NO
8160      278 FORMAT('NO',F7.4)
8170      WRITE (6,279) NO
8180      279 FORMAT('NO',F7.4)
8190      WRITE (6,280) NO
8200      280 FORMAT('NO',F7.4)
8210      WRITE (6,281) NO
8220      281 FORMAT('NO',F7.4)
8230      WRITE (6,282) NO
8240      282 FORMAT('NO',F7.4)
8250      WRITE (6,283) NO
8260      283 FORMAT('NO',F7.4)
8270      WRITE (6,284) NO
8280      284 FORMAT('NO',F7.4)
8290      WRITE (6,285) NO
8300      285 FORMAT('NO',F7.4)
8310      WRITE (6,286) NO
8320      286 FORMAT('NO',F7.4)
8330      WRITE (6,287) NO
8340      287 FORMAT('NO',F7.4)
8350      WRITE (6,288) NO
8360      288 FORMAT('NO',F7.4)
8370      WRITE (6,289) NO
8380      289 FORMAT('NO',F7.4)
8390      WRITE (6,290) NO
8400      290 FORMAT('NO',F7.4)
8410      WRITE (6,291) NO
8420      291 FORMAT('NO',F7.4)
8430      WRITE (6,292) NO
8440      292 FORMAT('NO',F7.4)
8450      WRITE (6,293) NO
8460      293 FORMAT('NO',F7.4)
8470      WRITE (6,294) NO
8480      294 FORMAT('NO',F7.4)
8490      WRITE (6,295) NO
8500      295 FORMAT('NO',F7.4)
8510      WRITE (6,296) NO
8520      296 FORMAT('NO',F7.4)
8530      WRITE (6,297) NO
8540      297 FORMAT('NO',F7.4)
8550      WRITE (6,298) NO
8560      298 FORMAT('NO',F7.4)
8570      WRITE (6,299) NO
8580      299 FORMAT('NO',F7.4)
8590      WRITE (6,300) NO
8600      300 FORMAT('NO',F7.4)
8610      WRITE (6,301) NO
8620      301 FORMAT('NO',F7.4)
8630      WRITE (6,302) NO
8640      302 FORMAT('NO',F7.4)
8650      WRITE (6,303) NO
8660      303 FORMAT('NO',F7.4)
8670      WRITE (6,304) NO
8680      304 FORMAT('NO',F7.4)
8690      WRITE (6,305) NO
8700      305 FORMAT('NO',F7.4)
8710      WRITE (6,306) NO
8720      306 FORMAT('NO',F7.4)
8730      WRITE (6,307) NO
8740      307 FORMAT('NO',F7.4)
8750      WRITE (6,308) NO
8760      308 FORMAT('NO',F7.4)
8770      WRITE (6,309) NO
8780      309 FORMAT('NO',F7.4)
8790      WRITE (6,310) NO
8800      310 FORMAT('NO',F7.4)
8810      WRITE (6,311) NO
8820      311 FORMAT('NO',F7.4)
8830      WRITE (6,312) NO
8840      312 FORMAT('NO',F7.4)
8850      WRITE (6,313) NO
8860      313 FORMAT('NO',F7.4)
8870      WRITE (6,314) NO
8880      314 FORMAT('NO',F7.4)
8890      WRITE (6,315) NO
8900      315 FORMAT('NO',F7.4)
8910      WRITE (6,316) NO
8920      316 FORMAT('NO',F7.4)
8930      WRITE (6,317) NO
8940      317 FORMAT('NO',F7.4)
8950      WRITE (6,318) NO
8960      318 FORMAT('NO',F7.4)
8970      WRITE (6,319) NO
8980      319 FORMAT('NO',F7.4)
8990      WRITE (6,320) NO
9000      320 FORMAT('NO',F7.4)
9010      WRITE (6,321) NO
9020      321 FORMAT('NO',F7.4)
9030      WRITE (6,322) NO
9040      322 FORMAT('NO',F7.4)
9050      WRITE (6,323) NO
9060      323 FORMAT('NO',F7.4)
9070      WRITE (6,324) NO
9080      324 FORMAT('NO',F7.4)
9090      WRITE (6,325) NO
9100      325 FORMAT('NO',F7.4)
9110      WRITE (6,326) NO
9120      326 FORMAT('NO',F7.4)
9130      WRITE (6,327) NO
9140      327 FORMAT('NO',F7.4)
9150      WRITE (6,328) NO
9160      328 FORMAT('NO',F7.4)
9170      WRITE (6,329) NO
9180      329 FORMAT('NO',F7.4)
9190      WRITE (6,330) NO
9200      330 FORMAT('NO',F7.4)
9210      WRITE (6,331) NO
9220      331 FORMAT('NO',F7.4)
9230      WRITE (6,332) NO
9240      332 FORMAT('NO',F7.4)
9250      WRITE (6,333) NO
9260      333 FORMAT('NO',F7.4)
9270      WRITE (6,334) NO
9280      334 FORMAT('NO',F7.4)
9290      WRITE (6,335) NO
9300      335 FORMAT('NO',F7.4)
9310      WRITE (6,336) NO
9320      336 FORMAT('NO',F7.4)
9330      WRITE (6,337) NO
9340      337 FORMAT('NO',F7.4)
9350      WRITE (6,338) NO
9360      338 FORMAT('NO',F7.4)
9370      WRITE (6,339) NO
9380      339 FORMAT('NO',F7.4)
9390      WRITE (6,340) NO
9400      340 FORMAT('NO',F7.4)
9410      WRITE (6,341) NO
9420      341 FORMAT('NO',F7.4)
9430      WRITE (6,342) NO
9440      342 FORMAT('NO',F7.4)
9450      WRITE (6,343) NO
9460      343 FORMAT('NO',F7.4)
9470      WRITE (6,344) NO
9480      344 FORMAT('NO',F7.4)
9490      WRITE (6,345) NO
9500      345 FORMAT('NO',F7.4)
9510      WRITE (6,346) NO
9520      346 FORMAT('NO',F7.4)
9530      WRITE (6,347) NO
9540      347 FORMAT('NO',F7.4)
9550      WRITE (6,348) NO
9560      348 FORMAT('NO',F7.4)
9570      WRITE (6,349) NO
9580      349 FORMAT('NO',F7.4)
9590      WRITE (6,350) NO
9600      350 FORMAT('NO',F7.4)
9610      WRITE (6,351) NO
9620      351 FORMAT('NO',F7.4)
9630      WRITE (6,352) NO
9640      352 FORMAT('NO',F7.4)
9650      WRITE (6,353) NO
9660      353 FORMAT('NO',F7.4)
9670      WRITE (6,354) NO
9680      354 FORMAT('NO',F7.4)
9690      WRITE (6,355) NO
9700      355 FORMAT('NO',F7.4)
9710      WRITE (6,356) NO
9720      356 FORMAT('NO',F7.4)
9730      WRITE (6,357) NO
9740      357 FORMAT('NO',F7.4)
9750      WRITE (6,358) NO
9760      358 FORMAT('NO',F7.4)
9770      WRITE (6,359) NO
9780      359 FORMAT('NO',F7.4)
9790      WRITE (6,360) NO
9800      360 FORMAT('NO',F7.4)
9810      WRITE (6,361) NO
9820      361 FORMAT('NO',F7.4)
9830      WRITE (6,362) NO
9840      362 FORMAT('NO',F7.4)
9850      WRITE (6,363) NO
9860      363 FORMAT('NO',F7.4)
9870      WRITE (6,364) NO
9880      364 FORMAT('NO',F7.4)
9890      WRITE (6,365) NO
9900      365 FORMAT('NO',F7.4)
9910      WRITE (6,366) NO
9920      366 FORMAT('NO',F7.4)
9930      WRITE (6,367) NO
9940      367 FORMAT('NO',F7.4)
9950      WRITE (6,368) NO
9960      368 FORMAT('NO',F7.4)
9970      WRITE (6,369) NO
9980      369 FORMAT('NO',F7.4)
9990      WRITE (6,370) NO
10000      370 FORMAT('NO',F7.4)

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3290 13 FORMAT(//,5X,'TORSIONAL PROPERTIES MATRIX',//)
3300 DO 14 I=1,NPGNFI
3310 14 WRITE(6,80) (IPSOLI(I,J),J=1,NPGYFI)
3320 705 CONTINUE
3330 WRITE(6,150) DIFF
3340 15 FORMAT(//,5X,'DIFFERENCE IN AREAS=',E11.4)
3350 WRITE(7,17) (NDM,IMAX,JYMAX,NPGNFI,NPGYFI,FX1,XC(1))
3360 WRITE(7,2) (XGFI(I),I=1,IMAX)
3370 WRITE(7,2) (YGF(J),J=1,JYMAX)
3380 DO 707 I=1,NPGNFI
3390 WRITE(7,80) (IPSOLI(I,J),J=1,NPGYFI)
3400 707 CONTINUE
3410 17 FORMAT(510,2F8.4)
3420 C
3430 C TORSIONAL CONSTANT CALCULATION USING FINITE DIFFERENCE
3440 ERR=.000001
3450 IMAXM1=IMAX-1
3460 JYMAXM1=JYMAX-1
3470 DO 710 I=2,IMAXM1
3480 DO 710 J=2,JYMAXM1
3490 IPSOL=1
3500 IF(IPSOLI(I,J).EQ.0.OR.IPSOLI(I+1,J).EQ.0.OR.
3510 * IPSOLI(I+1,J+1).EQ.0.OR.IPSOLI(I,J+1).EQ.0) IPSOL=0
3520 AIJ=-2.*(XGFI(I-1)-XGFI(I)).*(XGFI(I+1)-XGFI(I))
3530 AIJ=AIJ-2.*(YGF(J-1)-YGF(J)).*(YGF(J+1)-YGF(J))
3540 AIM1(I,J)=2.*(XGFI(I-1)-XGFI(I)).*(XGFI(I-1)-XGFI(I+1))/AIJ
3545 * +IPSOL
3550 AIP1(I,J)=-2.*(XGFI(I+1)-XGFI(I)).*(XGFI(I-1)-XGFI(I+1))/AIJ
3555 * +IPSOL
3560 AJM1(I,J)=2.*(YGF(J-1)-YGF(J)).*(YGF(J-1)-YGF(J+1))/AIJ
3565 * +IPSOL
3570 AJP1(I,J)=-2.*(YGF(J+1)-YGF(J)).*(YGF(J-1)-YGF(J+1))/AIJ
3575 * +IPSOL
3580 EPS(I,J)=2.*(AIJ+IPSOL)
3590 710 CONTINUE
3600 DO 720 I=1,IMAX
3610 DO 720 J=1,JYMAX
3620 720 F(I,J)=0.
3630 OMEG=1.
3640 OMO=1.-OMEG
3650 DO 740 IT=1,100
3660 AMAX=0.0
3670 DO 730 I=2,IMAX
3680 DO 730 J=2,JYMAX
3690 FOLD=F(I,J)
3700 FNEW=AIM1(I,J)*F(I-1,J)+AIP1(I,J)*F(I+1,J)
3710 FNEW=FNEW+AJM1(I,J)*F(I,J-1)+AJP1(I,J)*F(I,J+1)+EPS(I,J)
3720 F(I,J)=FNEW+OMEG+OMO*FOLD
3730 VAL=ABS(F(I,J)-FOLD)
3740 IF(VAL.GT.AMAX) AMAX=VAL
3750 730 CONTINUE
3760 IF(AMAX.LT.ERR) GO TO 750
3770 740 CONTINUE
3780 WRITE(6,740) IT
3790 745 FORMAT(5X,'ENCODES ITERATIONS',IT=,I4)
3800 750 CONTINUE
3810 C
3820 C INTEGRATION ROUTINE
3830 AJ=0
3840 DO 760 I=1,IMAXM1
3850 DO 760 J=1,JYMAXM1
3860 Z=F(I,J)+F(I+1,J)+F(I+1,J+1)+F(I,J+1)
3870 AJ=AJ+.25*Z*(XGFI(I+1)-XGFI(I))*(YGF(J+1)-YGF(J))
3880 760 CONTINUE
3890 AJ=2.*AJ
3900 AJ0=AJ
3910 WRITE(6,770) AJ0
3920 770 FORMAT(//,5X,'JTHETA=',E11.4//)
3930 * STOP
3940 END

```

APPENDIX E
MESH GENERATING PROGRAM

APPENDIX E

Mesh Generating Program

Program Description

The mesh generating program develops a two-dimensional, finite element mesh for circular ring cross sections of the type just discussed. It reads the nodal locations from a data file, assigns node numbers, determines element connectivity, computes applied forces and outputs a data file which can be used with the finite element code SAPIV (or other codes with modification).

The input data file is created by the section properties program (Appendix D). It contains information about the number of nodes in the x- and y- directions and the nodal coordinates. The cross section geometry is also in the data file. Thus, the mesh generating program reads the nodal coordinates and assigns node numbers to the nodes. It numbers in the shortest direction to minimize the stiffness matrix bandwidth. SAPIV requires the y-coordinates to be in the radial direction and the z-coordinate to be in the axial direction, so the program transforms the x-y-coordinates to the y-z-coordinate system.

Using the cross section geometry information, the program next develops the element connectivity. The node and element information is written to the output file.

Next, the applied loads data are input interactively. The program gives the choice of inputting loads at individual nodes or applying pressure loads on surfaces.

The program properly formats this information and writes it to the output file.

Program Input

Much of the program input is created by the section properties program and instructions for that program should be followed. After the mesh generating program reads the input file and assigns node and element numbers, it prints out node and element maps. These maps are then used for the following interactive input:

Constrained nodes: The program asks for the number of constrained nodes, the node numbers, and the directions in which they are constrained (y-z coordinates).

Applied loads: The program asks whether loads will be applied as concentrated loads to individual nodes (manual input) or as pressure loads to surfaces. For manual input, the program asks for the number of loaded nodes, the node numbers, the magnitude of the force and the direction (radial or axial) of the force. For pressure loading, the program asks for the applied pressure and the number of surfaces to which the pressure is applied. Then, for each surface, the program asks for the number of nodes on the surface, the direction in which the pressure acts and the node numbers on the surface.

A face pressure may also be calculated. This is the pressure on the sealing face and is assumed to vary linearly from the applied pressure to zero across the width of the face. Again,

the program asks for the number of nodes on the face, the direction in which the pressure acts, and the node numbers.

Sample Problem

The cross section shown in Figure E-1 is considered as a sample problem. It is a segment from an infinitely long cylinder which has an external pressure of 300 psi applied to it. This problem allows comparison between the numerical and theoretical stress distributions. The input file created by the section properties program is shown and the program run with interactive input underlined is shown next. The data file created by the program and the program listing are shown after that.

Results

The theoretical stress distribution of a pressurized thick-walled pressure vessel is ¹

$$\sigma_r = \frac{r_o^2 r_i^2}{r_o^2 - r_i^2} (p_o - p_i) \frac{1}{r^2} + \frac{p_i r_i^2 - p_o r_o^2}{r_o^2 - r_i^2}$$

$$\sigma_\theta = - \frac{r_o r_i^2}{r_o^2 - r_i^2} (p_o + p_i) \frac{1}{r^2} + \frac{p_i r_i^2 - p_o r_o^2}{r_o^2 - r_i^2}$$

where

σ_r = radial stress distribution

σ_θ = hoop (tangential) stress distribution

¹ Oden, J. T., Ripperrrger, E. A. Mechanics of Elastic Structures. McGraw-Hill, 1981, p. 91.

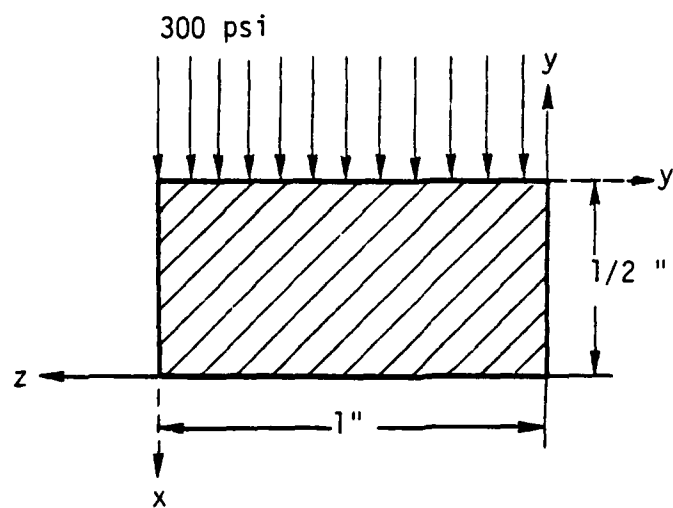


Figure E-1. Sample Cross Section.

r_o = outside radius (3.0 in)

r_i = inside radius (2.5 in)

r = radial distance

p_o = outside pressure (300 psi)

p_i = inside pressure (0. psi)

Using the data set created, the SAPIV program was run and comparisons of the theoretical and numerical stress distributions are shown in Figures E-2 and E-3.

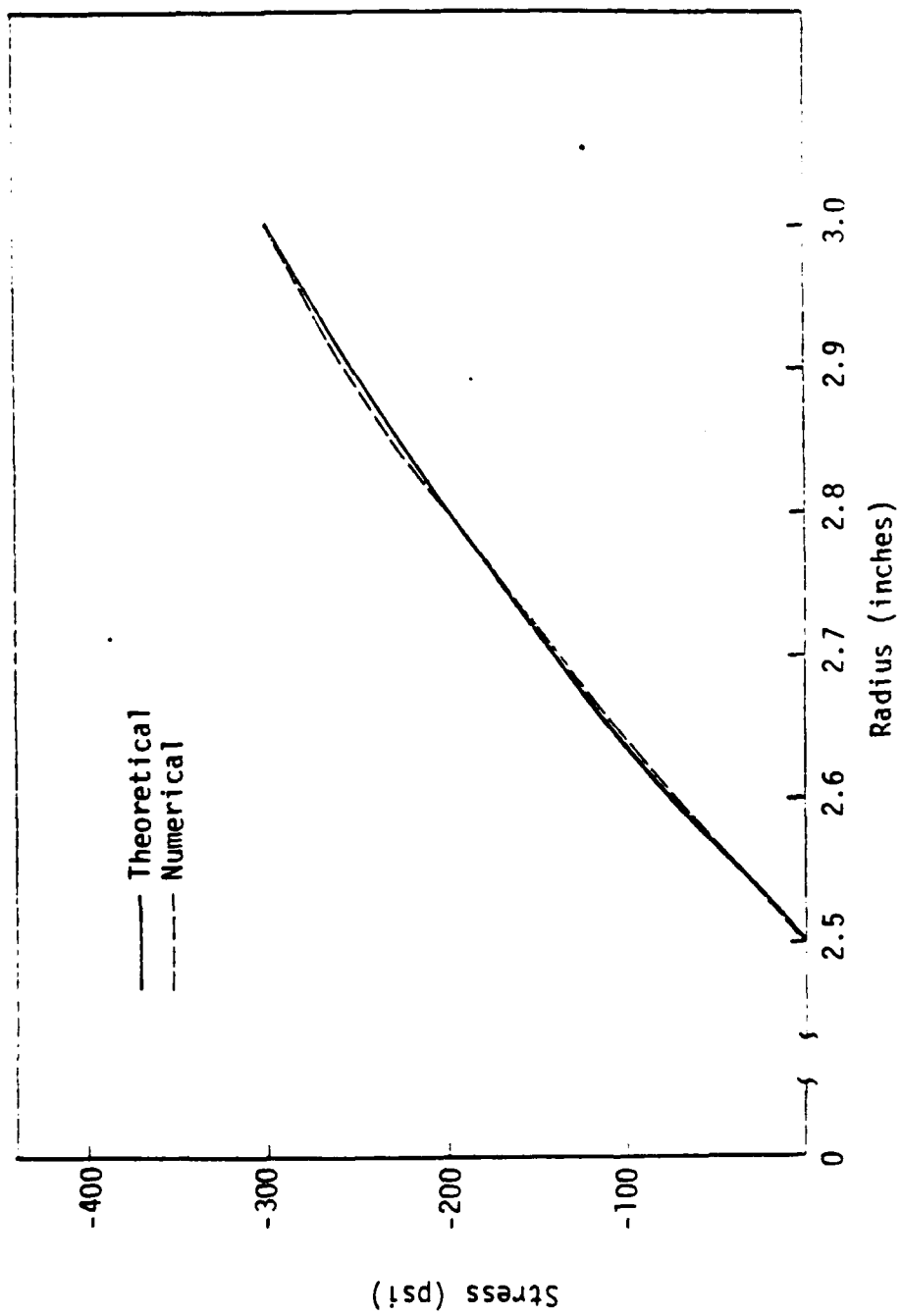


Figure E-2. Radial Stress Distribution (σ_r , S11)

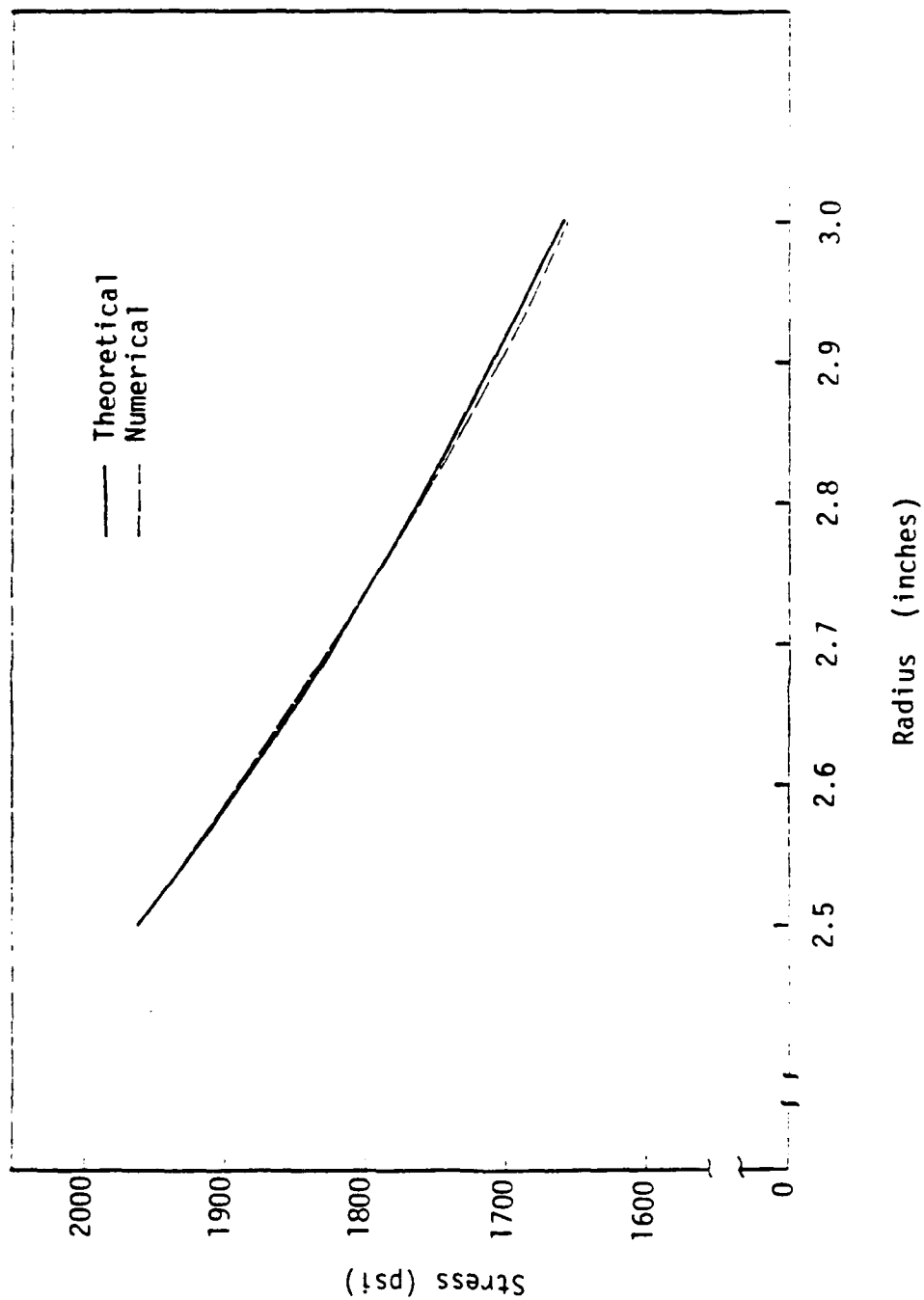


Figure E-3. Hoop Stress Distribution (σ_{θ} , 533)

Input File Created by the Section Properties Program

```

1 1 12 12 12 12 5.0000 0.
2 0. 0.0455 0.0909 0.1364 0.1818 0.2273 0.2727 0.3182 0.3636
3 0.4091 0.4545 0.5000
4 0. 0.0909 0.1818 0.2727 0.3636 0.4545 0.5455 0.6364 0.7273
5 0.9182 0.9091 1.0000
6 0 0 0 0 0 0 0 0 0 0 0
7 0 1 1 1 1 1 1 1 1 1 1 0
8 0 1 1 1 1 1 1 1 1 1 1 0
9 0 1 1 1 1 1 1 1 1 1 1 0
10 0 1 1 1 1 1 1 1 1 1 1 0
11 0 1 1 1 1 1 1 1 1 1 1 0
12 0 1 1 1 1 1 1 1 1 1 1 0
13 0 1 1 1 1 1 1 1 1 1 1 0
14 0 1 1 1 1 1 1 1 1 1 1 0
15 0 1 1 1 1 1 1 1 1 1 1 0
16 0 1 1 1 1 1 1 1 1 1 1 0
17 0 1 1 1 1 1 1 1 1 1 1 0
18 0 0 0 0 0 0 0 0 0 0 0 0

```

Program Run with Interactive Input

FEMESH 07/08/84 11.11:00

NODE MAP:

1	23	25	37	49	61	73	85	97	109	121	133
2	14	26	38	50	62	74	86	98	110	122	134
3	15	27	39	51	63	75	87	99	111	123	135
4	16	28	40	52	64	76	88	100	112	124	136
5	17	29	41	53	65	77	89	101	113	125	137
6	18	30	42	54	66	78	90	102	114	126	138
7	19	31	43	55	67	79	91	103	115	127	139
8	20	32	44	56	68	80	92	104	116	128	140
9	21	33	45	57	69	81	93	105	117	129	141
10	22	34	46	58	70	82	94	106	118	130	142
11	23	35	47	59	71	83	95	107	119	131	143
12	24	36	48	60	72	84	96	108	120	132	144

ELEMENT MAP:

1	12	23	34	45	56	67	78	89	100	111
2	13	24	35	46	57	68	79	90	101	112
3	14	25	36	47	58	69	80	91	102	113
4	15	26	37	48	59	70	81	92	103	114
5	16	27	38	49	60	71	82	93	104	115
6	17	28	39	50	61	72	83	94	105	116
7	18	29	40	51	62	73	84	95	106	117
8	19	30	41	52	63	74	85	96	107	118
9	20	31	42	53	64	75	86	97	108	119
10	21	32	43	54	65	76	87	98	109	120
11	22	33	44	55	66	77	88	99	110	121

HOW MANY NODES ARE CONSTRAINED?

01

INPUT FIXED NODES IN ASCENDING ORDER AND DIRECTION. 1=1, 2=2 BOTH=2

0000

DO YOU WANT TO MANUALLY INPUT LOADS? YES=1, NO=0

00

WHAT IS THE PRESSURE?

1000.0

HOW MANY SURFACES HAVE PRESSURE? EXCLUDE SEALING FACE

01

SURFACE # 1

HOW MANY NODES ON THIS SURFACE?

0000

WHICH DIRECTION IS PRESSURE ACTING?

1=1, 2=2, 3=3, 4=4

01

INPUT NODES ONE AT A TIME IN ORDER:

01
02
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144

IS A FREE SURFACE TO BE CALCULATED? YES=1, NO=0

0000

TIME 0.000000

Card Image Format Data File for SAPIV

HOOP STRESS CALCULATION									
NO.	1	2	3	4	5	6	7	8	9
1	1	1	0	0	0	0	0	0	0
2	1	0	1	1	1	1	1	0	0
3	1	0	1	1	1	1	1	0	0
4	1	0	1	1	1	1	1	0	0
5	1	0	1	1	1	1	1	0	0
6	1	0	1	1	1	1	1	0	0
7	1	0	1	1	1	1	1	0	0
8	1	0	1	1	1	1	1	0	0
9	1	0	1	1	1	1	1	0	0
10	1	0	1	1	1	1	1	0	0
11	1	0	1	1	1	1	1	0	0
12	1	0	1	1	1	1	1	0	0
13	1	0	1	1	1	1	1	0	0
14	1	0	1	1	1	1	1	0	0
15	1	0	1	1	1	1	1	0	0
16	1	0	1	1	1	1	1	0	0
17	1	0	1	1	1	1	1	0	0
18	1	0	1	1	1	1	1	0	0
19	1	0	1	1	1	1	1	0	0
20	1	0	1	1	1	1	1	0	0
21	1	0	1	1	1	1	1	0	0
22	1	0	1	1	1	1	1	0	0
23	1	0	1	1	1	1	1	0	0
24	1	0	1	1	1	1	1	0	0
25	1	0	1	1	1	1	1	0	0
26	1	0	1	1	1	1	1	0	0
27	1	0	1	1	1	1	1	0	0
28	1	0	1	1	1	1	1	0	0
29	1	0	1	1	1	1	1	0	0
30	1	0	1	1	1	1	1	0	0
31	1	0	1	1	1	1	1	0	0
32	1	0	1	1	1	1	1	0	0
33	1	0	1	1	1	1	1	0	0
34	1	0	1	1	1	1	1	0	0
35	1	0	1	1	1	1	1	0	0
36	1	0	1	1	1	1	1	0	0
37	1	0	1	1	1	1	1	0	0
38	1	0	1	1	1	1	1	0	0
39	1	0	1	1	1	1	1	0	0
40	1	0	1	1	1	1	1	0	0
41	1	0	1	1	1	1	1	0	0
42	1	0	1	1	1	1	1	0	0
43	1	0	1	1	1	1	1	0	0
44	1	0	1	1	1	1	1	0	0
45	1	0	1	1	1	1	1	0	0
46	1	0	1	1	1	1	1	0	0
47	1	0	1	1	1	1	1	0	0
48	1	0	1	1	1	1	1	0	0
49	1	0	1	1	1	1	1	0	0
50	1	0	1	1	1	1	1	0	0
51	1	0	1	1	1	1	1	0	0
52	1	0	1	1	1	1	1	0	0
53	1	0	1	1	1	1	1	0	0
54	1	0	1	1	1	1	1	0	0
55	1	0	1	1	1	1	1	0	0
56	1	0	1	1	1	1	1	0	0
57	1	0	1	1	1	1	1	0	0
58	1	0	1	1	1	1	1	0	0
59	1	0	1	1	1	1	1	0	0
60	1	0	1	1	1	1	1	0	0
61	1	0	1						

DATE	DESCRIPTION	AMOUNT	CHECK NO.	BANK	INTEREST	TOTAL
1994-01-01	Initial deposit	1000.00		Bank of America		1000.00
1994-01-15	Deposit	500.00	101	Bank of America		1500.00
1994-02-01	Withdrawal	250.00	102	Bank of America		1250.00
1994-02-15	Deposit	750.00	103	Bank of America		2000.00
1994-03-01	Withdrawal	100.00	104	Bank of America		1900.00
1994-03-15	Deposit	300.00	105	Bank of America		2200.00
1994-04-01	Withdrawal	150.00	106	Bank of America		2050.00
1994-04-15	Deposit	400.00	107	Bank of America		2450.00
1994-05-01	Withdrawal	200.00	108	Bank of America		2250.00
1994-05-15	Deposit	600.00	109	Bank of America		2850.00
1994-06-01	Withdrawal	100.00	110	Bank of America		2750.00
1994-06-15	Deposit	500.00	111	Bank of America		3250.00
1994-07-01	Withdrawal	300.00	112	Bank of America		2950.00
1994-07-15	Deposit	700.00	113	Bank of America		3650.00
1994-08-01	Withdrawal	150.00	114	Bank of America		3500.00
1994-08-15	Deposit	450.00	115	Bank of America		3950.00
1994-09-01	Withdrawal	250.00	116	Bank of America		3700.00
1994-09-15	Deposit	650.00	117	Bank of America		4350.00
1994-10-01	Withdrawal	100.00	118	Bank of America		4250.00
1994-10-15	Deposit	550.00	119	Bank of America		4800.00
1994-11-01	Withdrawal	350.00	120	Bank of America		4450.00
1994-11-15	Deposit	750.00	121	Bank of America		5200.00
1994-12-01	Withdrawal	150.00	122	Bank of America		5050.00
1994-12-15	Deposit	450.00	123	Bank of America		5500.00
1995-01-01	Withdrawal	250.00	124	Bank of America		5250.00
1995-01-15	Deposit	650.00	125	Bank of America		5900.00
1995-02-01	Withdrawal	100.00	126	Bank of America		5800.00
1995-02-15	Deposit	500.00	127	Bank of America		6300.00
1995-03-01	Withdrawal	300.00	128	Bank of America		6000.00
1995-03-15	Deposit	700.00	129	Bank of America		6700.00
1995-04-01	Withdrawal	150.00	130	Bank of America		6550.00
1995-04-15	Deposit	450.00	131	Bank of America		7000.00
1995-05-01	Withdrawal	250.00	132	Bank of America		6750.00
1995-05-15	Deposit	650.00	133	Bank of America		7400.00
1995-06-01	Withdrawal	100.00	134	Bank of America		7300.00
1995-06-15	Deposit	500.00	135	Bank of America		7800.00
1995-07-01	Withdrawal	300.00	136	Bank of America		7500.00
1995-07-15	Deposit	700.00	137	Bank of America		8200.00
1995-08-01	Withdrawal	150.00	138	Bank of America		8050.00
1995-08-15	Deposit	450.00	139	Bank of America		8500.00
1995-09-01	Withdrawal	250.00	140	Bank of America		8250.00
1995-09-15	Deposit	650.00	141	Bank of America		8900.00
1995-10-01	Withdrawal	100.00	142	Bank of America		8800.00
1995-10-15	Deposit	500.00	143	Bank of America		9300.00
1995-11-01	Withdrawal	300.00	144	Bank of America		9000.00
1995-11-15	Deposit	700.00	145	Bank of America		9700.00
1995-12-01	Withdrawal	150.00	146	Bank of America		9550.00
1995-12-15	Deposit	450.00	147	Bank of America		10000.00
1996-01-01	Withdrawal	250.00	148	Bank of America		9750.00
1996-01-15	Deposit	650.00	149	Bank of America		10400.00
1996-02-01	Withdrawal	100.00	150	Bank of America		10300.00
1996-02-15						

281	107	129	128	116	117	1	0	0	20	1	1.0
282	108	130	129	117	118	1	0	0	20	1	1.0
283	109	131	130	118	119	1	0	0	20	1	1.0
284	110	132	131	119	120	1	0	0	20	1	1.0
285	111	133	132	120	121	1	0	0	20	1	1.0
286	112	134	133	121	122	1	0	0	20	1	1.0
287	113	135	134	122	123	1	0	0	20	1	1.0
288	114	136	135	123	124	1	0	0	20	1	1.0
289	115	137	136	124	125	1	0	0	20	1	1.0
290	116	138	137	125	126	1	0	0	20	1	1.0
291	117	139	138	126	127	1	0	0	20	1	1.0
292	118	140	139	127	128	1	0	0	20	1	1.0
293	119	141	140	128	129	1	0	0	20	1	1.0
294	120	142	141	129	130	1	0	0	20	1	1.0
295	121	143	142	130	131	1	0	0	20	1	1.0
296	122	144	143	131	132	1	0	0	20	1	1.0
297	1	1	0	-40.9050	0	0	0	0	0	0	0
298	13	1	0	-81.8100	0	0	0	0	0	0	0
299	25	1	0	-81.8100	0	0	0	0	0	0	0
300	37	1	0	-81.8100	0	0	0	0	0	0	0
301	49	1	0	-81.8100	0	0	0	0	0	0	0
302	61	1	0	-81.8550	0	0	0	0	0	0	0
303	73	1	0	-81.8550	0	0	0	0	0	0	0
304	85	1	0	-81.8100	0	0	0	0	0	0	0
305	97	1	0	-81.8100	0	0	0	0	0	0	0
306	109	1	0	-81.8100	0	0	0	0	0	0	0
307	121	1	0	-81.8100	0	0	0	0	0	0	0
308	133	1	0	-40.9050	0	0	0	0	0	0	0
309		1.0		1.0		1.0		1.0			
310											
311											
312											

```

10 C MESH GENERATING PROGRAM
20 C
30 C THIS PROGRAM GENERATES A 2-D FINITE ELEMENT MESH FOR A
40 C GIVEN CROSS SECTION. THE SECTION GEOMETRY IS READ FROM
50 C AN INPUT FILE CREATED BY THE "SECTION" PROGRAM. LOAD AND
60 C CONSTRAINT INFORMATION IS INPUT INTERACTIVELY. THE PROGRAM
70 C OUTPUT IS A CARD IMAGE FORMAT DATA FILE WHICH CAN BE USED
80 C WITH THE FINITE ELEMENT CODE "SAFIV."
90 C
100 DIMENSION IP(40,40), XPE(200), YPE(200), XGP(40), YGP(40)
110 DIMENSION IPSOLI(40,40), IELEM(40,40), NFINEL(20), IPN(20)
120 DIMENSION INK(200), FN(200), FEN(200)
130 CALL OPSYS(ALLOC, PEDATA, 7)
140 1 FORMAT(5D, 2F8.4)
150 2 FORMAT(5X, 2F7.4)
160 3 FORMAT(5X, 4D12)
170 4 FORMAT(20I4)
180 5 FORMAT(5X, 1D, 5X, F7.4, 5X, F7.4)
190 6 FORMAT(7X, 5X, NODE MAP(1,1))
200 7 FORMAT(7X, 5X, ELEMENT MAP(1,1))
210 8 FORMAT(7X, 5X, ELEMENT NUMBER(5X, 1D, 5X, F7.4, 5X, F7.4, 1D)
220 9 FORMAT(5I5, 2F10.4, 2I5, F10.1)
230 C
240 C READ SECTION GEOMETRY FROM INPUT FILE "PEDATA"
250 READ(7,1) NDM, IXMAX, JYMAX, NPGNF1, NPGYF1, RX1, XC1
260 DO 775 I=1, IXMAX
270 DO 775 J=1, JYMAX
280 775 IF(I, J)=0
290 READ(7,2) XGP(I, J), YGP(I, J), IELEM(I, J)
300 READ(7,3) XPSOLI(I, J), YPSOLI(I, J), IELEM(I, J)
310 DO 777 I=1, NPGNF1
320 777 READ(7,4) (IPSOLI(I, J), J=1, NPGYF1)
330 C
340 C NUMBER NODES - NUMBER IN SHORTEST DIRECTION TO MINIMIZE
350 C BANDWIDTH
360 YTOT=YGP(JYMAX)-YGP(1)
370 XTOT=XGP(IXMAX)-XGP(1)
380 IF(XTOT.LT.YTOT) GO TO 390
390 DO 300 I=2, NPGNF1
400 DO 300 J=2, NPGYF1
410 IF(IPSOLI(I, J).EQ.1) NEL=NEL+1
420 NEMAX=NEL
430 IF(IPSOLI(I, J).EQ.1) IELEM(I, J)=NEL
440 IF(IPSOLI(I, J).EQ.1 OR IPSOLI(I, J-1).EQ.1 OR IPSOLI(I-1, J).EQ.1)
450 300 300 IF(IPSOLI(I-1, J).EQ.1) GO TO 780
460 GO TO 300
470 780 INODE=INODE+1
480 IF(I-1, J-1).EQ.0)
490 780 INODE =-TOT+OF(I-1)
500 YFE INODE=RX1-XC1-XGP(I-1)
510 INODE=INODE
520 CONTINUE
530 GO TO 300
540 300 DO 320 J=2, NPGYF1
550 DO 320 I=2, NPGNF1
560 IF(IPSOLI(I, J).EQ.1) NEL=NEL+1
570 NEMAX=NEL
580 IF(IPSOLI(I, J).EQ.1) IELEM(I, J)=NEL
590 IF(IPSOLI(I, J).EQ.1 OR IPSOLI(I, J-1).EQ.1 OR IPSOLI(I-1, J).EQ.1)
600 320 320 IF(IPSOLI(I-1, J).EQ.1) GO TO 310
610 GO TO 320

```

```

610 810 INODE=INODE-1
620 IF(I-1,J-1)=INODE
630 ZFE(INODE)=MTOT-YSP(I-1)
640 YFE(INODE)=FNL+XCL-NSP(I-1)
650 MNODE=INODE
670 920 CONTINUE
680 910 CONTINUE
690 C
700 C PRINT NODE AND ELEMENT MAPS
710 WRITE(6,6)
720 DO 950 I=1,IXMAX
730 950 WRITE(6,4) (IF(I,J),J=1,JYMAX)
740 WRITE(6,7)
750 DO 955 I=2,IXMAX
760 955 WRITE(6,4) (IELEM(I,J),J=2,JYMAX)
770 C
780 C DEVELOP OUTPUT FILE
790 CALL OPSYS('ALLSC', 'DATA1',7)
800 C HEADING CARD
810 WRITE(7,20)
820 20 FORMAT(1X,'**** HOOF STRESS CALCULATION ****')
830 C MASTER CONTROL CARD
840 NUMNP=MNODE
850 NELTYP=1
860 LL=1
870 NF=0
880 NDYN=0
890 MODEX=0
900 NAD=0
910 KEGB=0
920 WRITE(7,30) NUMNP, NELTYP, LL, NF, NDYN, MODEX, NAD, KEGB
930 30 FORMAT(8I5)
940 C LOCAL POINT DATA
950 10 WRITE(8,10)
960 10 FORMAT(1X,5X,'HOW MANY NODES ARE CONSTRAINED?')
970 READ(5,*) NCON
980 WRITE(8,21)
990 21 FORMAT(1X,5X,'INPUT FIXED NODES IN ASCENDING ORDER ... AND ')
1000 10 DIRECTION. (Z=1,Y=2,BOTH=3)
1010 DO 17 I=1,NCON
1020 17 READ(5,*) NFIXED(I),IFD(I)
1030 INT=1
1040 IYT=0
1050 IZT=0
1060 INR=1
1070 IYR=1
1080 IZR=1
1090 AN=0
1100 T=0.0
1110 S=0.0
1120 DO 15 INODE=1,MNODE
1130 NGC=0
1140 DO 16 I=1,NCON
1150 IF(INODE.EQ.NFIXED(I)) NFN=1
1160 16 IF(INODE.EQ.NFIXED(I)) NGC=1
1170 IF(NGC.EQ.1) GO TO 14
1180 WRITE(7,32) INODE,INT,IYT,IZT,INR,IYR,IZR,X,YFE(INODE)
1190 32 FORMAT(1X,14,8I5,3F10,4,15,F5,1)
1200 GO TO 15
1210 14 IZT=1
1220 IYT=1
1230 IF(IFD(NFN).EQ.1) GO TO 15
1240 IF(IFD(NFN).NE.1) IZT=0
1250 IF(IFD(NFN).NE.2) IYT=0
1260 15 WRITE(7,32) INODE,INT,IYT,IZT,INR,IYR,IZR,X,YFE(INODE)

```

```

1280             , DFE(INODE), KN, T
1290             IET=0
1300             IYT=0
1310             35 CONTINUE
1320 C ELEMENT DATA
1330             NELT=4
1340             MNTC=1
1350             NTYA=0
1360             WRITE(7,40) NELT, NEMAX, NDM, MNTC, NTYA
1370             40 FORMAT(6I5)
1380 C MATERIAL PROPERTY INFORMATION
1390 C EACH MATERIAL MUST BE CONSIDERED
1400 C MATERIAL #1
1410             MIDN=1
1420             NDT=1
1430             WD=0.0
1440             AMD=0.0
1450             BETA=0.0
1460             WRITE(7,42) MIDN, NDT, WD, AMD, BETA
1470             42 FORMAT(2I5, 3F10.1)
1480             TEMP=0.0
1490             E=3.1E+06
1500             ANUS= 2
1510             G=E/2/(1+ANUS)
1520             WRITE(7,45) TEMP, E, E, E, ANUS, ANUS, ANUS, G
1530             45 FORMAT(8F10.1)
1540             CTE=0.000000
1550             WRITE(7,46) CTE, CTE, CTE
1560             46 FORMAT(3F10.6)
1570             IF (NDM.EQ.1) GO TO 70
1580 C MATERIAL #2
1590             MIDN=2
1600             WRITE(7,42) MIDN, NDT, WD, AMD, BETA
1610             E=3.1E+06
1620             ANUS= 2
1630             G=E/2/(1+ANUS)
1640             WRITE(7,45) TEMP, E, E, E, ANUS, ANUS, ANUS, G
1650             WRITE(7,46) CTE, CTE, CTE
1660             70 CONTINUE
1670 C ELEMENT LOAD FACTORS (FOUR CARDS)
1680             FTL=1
1690             FPL=1
1700             FG=0
1710             WRITE(7,47) FTL, FPL, FG, FG, FG
1720             47 FORMAT(5F10.1)
1730             WRITE(7,47) FG, FG, FG, FG, FG
1740             WRITE(7,47) FG, FG, FG, FG, FG
1750             WRITE(7,47) FG, FG, FG, FG, FG
1760 C ELEMENT NUMBER INFORMATION
1770             NEL=0
1780             RT=0.0
1790             FN=0.0
1800             NSQ=20
1810             K=1
1820             ET=1
1830             IF(YTOT, LT, XTOT) GO TO 50
1840             DO 957 J=1, JYMAX
1850                 DO 957 I=1, IXMAX
1860                     IF(IELEM(I, J).EQ.0) GO TO 957
1870                     NEL=NEL+1
1880                     II=IP(I, J)
1890                     JJ=IP(I-1, J)
1900                     KK=IP(I-1, J-1)
1910                     LL=IP(I, J-1)
1920                     WRITE(7,9) NEL, II, JJ, KK, LL, IPSOLI(I, J), RT, FN, NSQ, K, ET
1930             957 CONTINUE

```



```

1940      GOTO 60
1950      50 DO 55 I=1, IXMAX
1960          DO 55 J=1, JYMAX
1970              IF (IELEM(I, J).EQ. 0) GO TO 55
1980              NEL=NEL+1
1990              II=IP(I, J)
2000              JJ=IP(I-1, J)
2010              KK=IP(I-1, J-1)
2020              LL=IP(I, J-1)
2030              WRITE(7, 9) NEL, II, JJ, KK, LL, IPSOLI(I, J), RT, FN, NSO, K, ET
2040          55 CONTINUE
2050      60 CONTINUE
2060 C CONCENTRATED LOAD/MASS DATA
2070      DO 62 I=1, MNODE
2080          FY(I)=0.0
2090          FZ(I)=0.0
2100      62 CONTINUE
2110          WRITE(6, 61)
2120      61 FORMAT(/, 5X, 'DO YOU WANT TO MANUALLY INPUT LOADS? YES=1, '
2130          *, ' NO=0')
2140          READ(5, *) NA
2150          IF (NA.EQ. 1) GO TO 126
2160          WRITE(6, 79)
2170      79 FORMAT(/, 5X, 'WHAT IS THE PRESSURE?')
2180          READ(5, *) PW
2190          WRITE(6, 80)
2200      80 FORMAT(/, 5X, 'HOW MANY SURFACES HAVE PRESSURE? EXCLUDE '
2210          *, ' SEALING FACE')
2220          READ(5, *) NSURF
2230          DO 100 NS=1, NSURF
2240      91      WRITE(6, 83) NS
2250      93      FORMAT(/, 5X, 'SURFACE #', I2)
2260          WRITE(6, 85)
2270          IF (NS.EQ. 1) WRITE(6, 86)
2280          READ(5, *) NNSF
2290          WRITE(6, 87)
2300      87      FORMAT(/, 5X, 'WHICH DIRECTION IS PRESSURE ACTING? /, 5X, '
2310          *, ' -Y=1, +Y=2, -Z=3, +Z=4')
2320          READ(5, *) IDIR
2330          WRITE(6, 88)
2340      88      FORMAT(/, 5X, 'INPUT NODES (ONE AT A TIME IN ORDER) ')
2350          IF (IDIR.EQ. 3, OR, IDIR.EQ. 4) GO TO 95
2360 C SURFACE FORCES IN THE RADIAL DIRECTION
2370          DO 89 I=1, NNSF
2380      89      READ(5, *) IN(I)
2390          NM1=NNSF-1
2400          FAC=1.
2410          IF (IDIR.EQ. 1) FAC=-1.
2420          DO 90 I=1, NM1
2430              DZ=ABS(ZFE(IN(I+1))-ZFE(IN(I)))
2440              ALORD=PW*DZ*YFE(IN(I))
2450              FY(IN(I))=FY(IN(I))+ALORD/2.0*FAC
2460              FY(IN(I+1))=FY(IN(I+1))-ALORD/2.0*FAC
2470      90      CONTINUE
2480          GO TO 100
2490 C SURFACE FORCES IN THE AXIAL DIRECTION
2500      95      DO 96 I=1, NNSF
2510      96      READ(5, *) IN(I)
2520          NM1=NNSF-1
2530          FAC=1.
2540          IF (IDIR.EQ. 3) FAC=-1.
2550          DO 97 I=1, NM1
2560              ALORD=PW*ABS(YFE(IN(I+1))*2-YFE(IN(I))*2)/2.0
2570              FZ(IN(I))=FZ(IN(I))+ALORD/2.0*FAC
2580              FZ(IN(I+1))=FZ(IN(I+1))-ALORD/2.0*FAC
2590      97      CONTINUE

```

```

2600 100 CONTINUE
2610 C FACE LOADS
2620 WRITE(6,110)
2630 110 FORMAT(/,5X,'IS A FACE PRESSURE TO BE CALCULATED? YES=1'
2640 *,',NO=0')
2650 READ(5,*) NA
2660 IF(NA.EQ.0) GO TO 130
2670 WRITE(6,85)
2680 READ(5,*) NNSF
2690 WRITE(6,87)
2700 READ(5,*) IDIR
2710 WRITE(6,115)
2720 115 FORMAT(/,5X,'INPUT NODES FROM INSIDE RADIUS TO OUTSIDE')
2730 DO 120 I=1,NNSF
2740 120 READ(5,*) IN(I)
2750 NM1=NNSF-1
2760 FAC=1.
2770 IF(IDIR.EQ.3) FAC=-1.
2780 SLOPE=FW/(YFE(IN(NNSF))-YFE(1))
2790 B=FW-SLOPE*YFE(IN(NNSF))
2800 DO 125 I=1,NM1
2810 PRESS=SLOPE*(YFE(IN(I))+YFE(IN(I+1)))/2.0+B
2820 ALOAD=PRESS*(YFE(IN(I+1))**2-YFE(IN(I))**2)/2.0
2830 FZ(IN(I))=FZ(IN(I))+ALOAD/2.0*FAC
2840 FZ(IN(I+1))=FZ(IN(I+1))+ALOAD/2.0*FAC
2850 125 CONTINUE
2860 GO TO 130
2870 C MANUAL LOAD INPUT
2880 126 WRITE(6,127)
2890 127 FORMAT(/,5X,'HOW MANY NODES?')
2900 READ(5,*) INOD
2910 WRITE(6,128)
2920 128 FORMAT(/,5X,'INPUT NODE, LOAD, AND IS IT RADIAL(1) OR '
2930 *,',AXIAL(2)?')
2940 DO 129 I=1,INOD
2950 READ(5,*) N,F,IFA
2960 IF(IFA.EQ.1) FY(N)=F
2970 IF(IFA.EQ.2) FZ(N)=F
2980 129 CONTINUE
2990 130 CONTINUE
3000 NSLC=1
3010 FX=0.0
3020 AMX=0.0
3030 AMY=0.0
3040 AMZ=0.0
3050 DO 140 I=1,MNODE
3060 IF(FY(I).EQ.0. AND FZ(I).EQ.0.) GO TO 140
3070 WRITE(7,135) I,NSLC,FX,FY(I),FZ(I),AMX,AMY,AMZ
3080 135 FORMAT(215,6F10.4)
3090 140 CONTINUE
3100 WRITE(7,*)
3110 C ELEMENT LOAD MULTIPLIERS
3120 EM=1.
3130 WRITE(7,145) EM,EM,EM,EM
3140 145 FORMAT(4F10.1)
3150 WRITE(7,150)
3160 150 FORMAT(//,////)
3170 STOP
3180 END

```

APPENDIX F
HEAT TRANSFER ANALYSIS

APPENDIX F

Heat Transfer Analysis

Program Description

The heat transfer analysis program solves for the temperature distribution and thermal rotation for two materials rubbing together in a seal configuration. Data similar to that for the Section Properties program is needed as input to define the seal. Heat generation at the sliding interface arises from friction due to the mechanical contact. Boundary conditions around the seal are convective having an arbitrary convection coefficient. Once the temperature field is established, thermal rotation of the two rings is computed.

Data Input

With reference to the example problem shown in Figure F-1 and the following program listing, the following data input is needed:

XC, YC	x and y coordinates of the corner points input clockwise
RX1	Radius at corner point 1
NC	Number of corner points
DXMAX, DYMAX	Maximum grid spacing in x and y directions
XMIN1, YMIN1	Minimum coordinates of material 2
XMAX1, YMAX1	Maximum coordinates of material 2
	Material 2 must be a rectangular solid contained with the original boundary
COND (1 & 2)	Thermal conductivity of materials 1 and 2

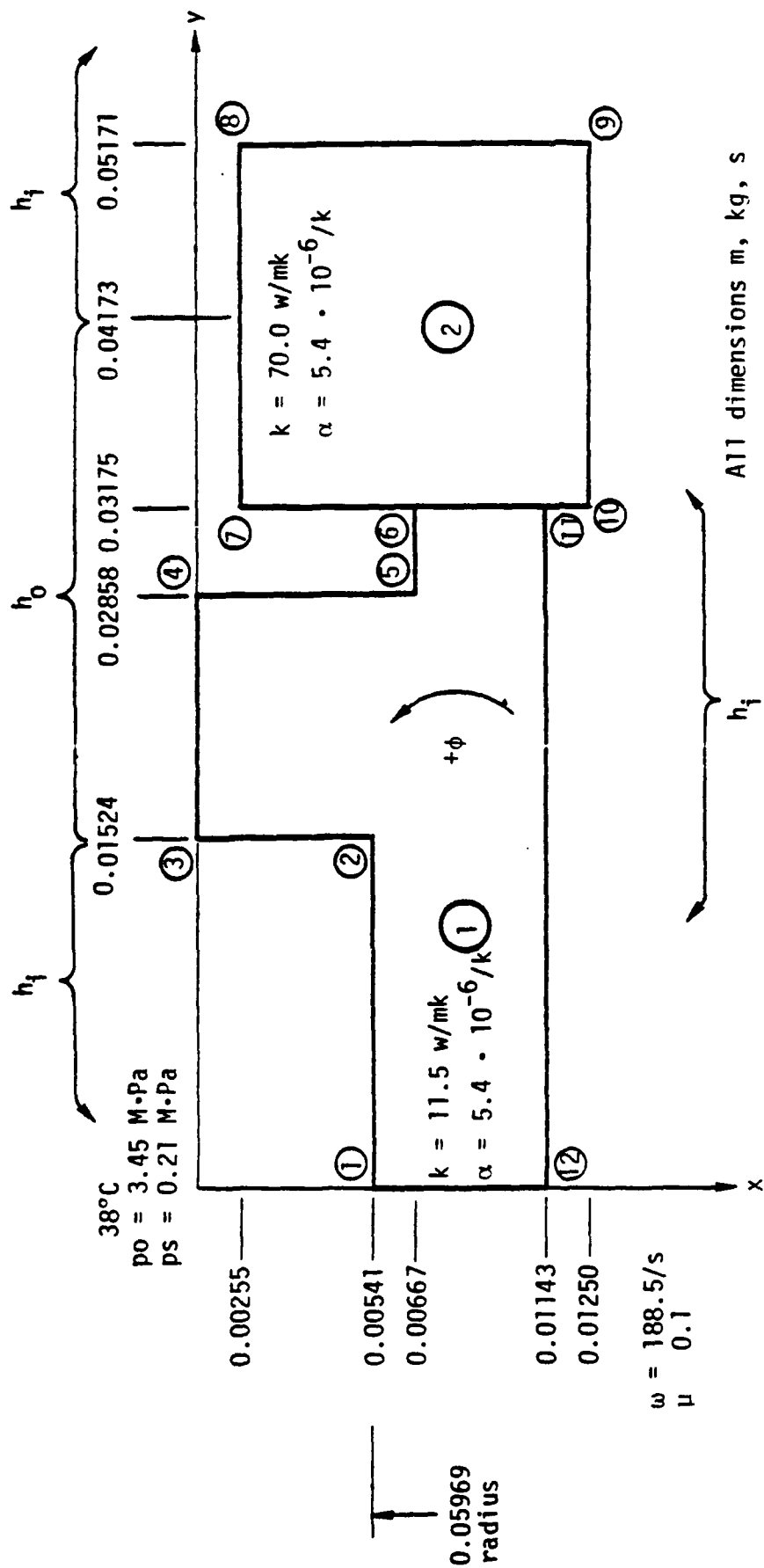


Figure F-1. Sample heat Transfer Problem.

$$h_o = 17700 \text{ w/m}^2\text{k}$$

$$h_i = 5900$$

ALPHA1 & ALPHA2 Thermal coefficient of expansion of materials 1 and 2

OMEG angular velocity

RO Outside radius of seal face contact

RI Inside radius of seal face contact

BAL Balance ratio of seal

PSP Spring load on face

PO Sealed pressure (outside pressure)

AMU Friction coefficient associated with mechanical contact pressure only

Q Computed heat flow into seal faces

TINF Seal environment temperature

HCM(I) Convection coefficient just behind and just ahead of corner point I. If the convection coefficient does not change at the corner point, then HCM = HCP. A change may be introduced as well. (See later example.)

XCHG(I) If there is to be a change in convection coefficient along the boundary between XC(I) and YCHGII) XC(I+1) or YC(I) and YC(I+1), then either XCHG(I) or YCHG(I) (as necessary) must be specified. The program knows when to expect such a change because there will be a difference between HCP(I) and HCM(I+1) in such cases. The program introduces a step change in convection coefficient at XCHG or

YCHG according to the values HCP(I) and HCM(I+1).

Proper interpolation between element spaces is done.

Example

Figure F-1 shows a seal used as an example. The data is input into the program as described. The output follows the program listing.

In the example, material (1) is carbon and material (2) is WC. The seal used for this example is the first wavy seal used for early research on waves in water [4]. Note how material (2) is included in the original boundary. The convection coefficient is changed along the (7) - (8) boundary.

The output from this program is shown. The refined mesh suggests the shape of the problem. Finally, the temperature distribution is shown. The direction of printing was changed 90° here to accommodate the longer than thick seal geometry. The -0 temperature values are meaningless temperature values at locations where there is no material. The -0 represents a roundoff for zero from the negative side. Thermal rotations are also shown using the sign conventions shown in Figure F-1.

*Figure F-1 will be provided
later.*

```

10 C PROGRAM TO CALCULATE SECTION PROPERTIES, HEAT TRANSFER, AND
20 C THERMAL ROTATION FOR ARBITRARY CROSS SECTIONS MADE UP OF ANY
30 C NUMBER OF STRAIGHT LINE SEGMENTS PARALLEL TO THE X AND Y AXES.
40 C
50 C INPUT NUMBER OF POINTS NC AND THE POINTS X(1), Y(1) IN
60 C CLOCKWISE DIRECTION AROUND THE PART. STARTING POINT IS
70 C ARBITRARY. EACH POINT BY DEFINITION MUST BE A CORNER.
80 C THIS PROGRAM ACCEPTS A MATERIAL #1 IN A RECTANGULAR SHAPE ONLY.
90 C THE XMIN,YMIN AND XMAX,YMAX CORNERS OF THE MATERIAL MUST BE
100 C SPECIFIED. MATERIAL #2 MUST BE WITHIN THE ORIGINAL BOUNDARY.
110 C THIS PROGRAM CONSIDERS THE VERTICAL (Y=CONST) INTERFACE BETWEEN
120 C THE TWO MATERIALS TO BE A UNIFORM HEAT SOURCE Q WMM*2 TO SIMULATE
130 C A SEAL INTERFACE.
140 C SI UNITS M,KG,S ARE USE
150 C MATERIAL 1 IS CARBON, MATERIAL 2 IS NO
160 C DIMENSION NG(22),YG(22),NGF(48),YGF(48),XG(22),YG(22)
170 C DIMENSION AIM1(48,48),AIP1(48,48),AJM1(48,48),AJP1(48,48)
180 C DIMENSION T(48,48),S1J(48,48),HGM(22),HGF(22),RSP(48)
190 C DIMENSION IEX(22),IGX(22),JGYC(22),ISOLID(22,12),IGN(48)
200 C DIMENSION JSN(48),IFBOLI(48,48),IS(4)
210 C DIMENSION XCHG(22),XCHG(22),COND(2)
220 C DIMENSION TTT(30)
230 C CALL OPENSCAL(30,'DATA1',T)
240 C REAL*4 X(1),X(2),Y(1),Y(2),X(3),X(4),Y(3),Y(4)
250 C REAL*4 XIP1,XIM1,XIP1,XIM1
260 C DATA NG(1),NG(2),NG(3),NG(4),NG(5),NG(6),NG(7),NG(8),NG(9),NG(10),
270 C 1,NG(12),NG(13),NG(14),NG(15),NG(16),NG(17),NG(18),NG(19),NG(20),
280 C 1,NG(22),NG(23),NG(24),NG(25),NG(26),NG(27),NG(28),NG(29),NG(30)
290 C INPUT MATERIAL #2 COORDINATES XMIN,YMIN,XMAX,YMAX
300 C IF THE MATERIALS ARE NOT USED THEN SET NG(1) TO NG(30) TO 0
310 C XMIN1=0.00255
320 C YMIN1=0.00175
330 C XMAX1=0.01250
340 C YMAX1=0.05171
350 C INPUT RADIUS AT X1
360 C RX1=0.05969
370 C NC=12
380 C NG=NC/2
390 C INPUT GRID SIZE FOR FINER MESH
400 C DXMAX=0.0015
410 C DYMAX=0.0015
420 C PIS=0.14159265
430 C NG=17700
440 C RI=NC/2
450 C JIM=1,ELL=7
460 C JIM=1,ELL=7
470 C RI=0.15-0.05
480 C RI=0.15-0.05
490 C JIM=1,ELL=7
500 C RI=0.04826
510 C RI=0.04826
520 C RI=0.04826
530 C RI=0.04826
540 C RI=0.04826
550 C RI=0.04826
560 C RI=0.04826
570 C RI=0.04826
580 C RI=0.04826
590 C RI=0.04826
600 C RI=0.04826
610 C RI=0.04826
620 C RI=0.04826
630 C RI=0.04826
640 C RI=0.04826
650 C RI=0.04826
660 C RI=0.04826
670 C RI=0.04826
680 C RI=0.04826
690 C RI=0.04826
700 C RI=0.04826
710 C RI=0.04826
720 C RI=0.04826
730 C RI=0.04826
740 C RI=0.04826
750 C RI=0.04826
760 C RI=0.04826
770 C RI=0.04826
780 C RI=0.04826
790 C RI=0.04826
800 C RI=0.04826
810 C RI=0.04826
820 C RI=0.04826
830 C RI=0.04826
840 C RI=0.04826
850 C RI=0.04826
860 C RI=0.04826
870 C RI=0.04826
880 C RI=0.04826
890 C RI=0.04826
900 C RI=0.04826
910 C RI=0.04826
920 C RI=0.04826
930 C RI=0.04826
940 C RI=0.04826
950 C RI=0.04826
960 C RI=0.04826
970 C RI=0.04826
980 C RI=0.04826
990 C RI=0.04826

```



```

870      TINF=38.
880 C ENTER THE POINTS ALONG THE BOUNDARY WHERE CONVECTION COEFFICIENT
890 C CHANGES
900      DO 21 I=1,NC
910          VCHG(I)=0.
920 21      XCHG(I)=0.
930 C ENTER THE CONVECTION COEFFICIENTS
940      HCM(1)=HI
950      HCP(1)=HI
960      HCM(2)=HI
970      HCP(2)=HI
980      HCM(3)=HI
990      HCP(3)=HO
1000     HCM(4)=HO
1010     HCP(4)=HO
1020     HCM(5)=HO
1030     HCP(5)=HO
1040     HCM(6)=HO
1050     HCP(6)=HO
1060     HCM(7)=HO
1070     HCP(7)=HO
1080     HCM(8)=HI
1090     HCP(8)=HI
1100     HCM(9)=HI
1110     HCP(9)=HI
1120     HCM(10)=HI
1130     HCP(10)=HI
1140     HCM(11)=HI
1150     HCP(11)=HI
1160     HCM(12)=HI
1170     HCP(12)=HI
1180     VCHG(7)=0.34175
1190 C SET UP THE GRID POINTS
1200     DO 20 I=1,NG
1210         NG(I)=NG(I)+1
1220     20 V(I)=V(I)+1.
1230 C ARRANGE THE GRID POINTS IN INCREASING ORDER
1240     DO 22 I=1,NG
1250         IST=I
1260         AMIN=1.E+36
1270         DO 22 J=IST,NG
1280             IF(VG(J).GT.AMIN) GO TO 22
1290             AMIN=VG(J)
1300             J=1
1310         22 CONTINUE
1320         YTMP=NG(IST)
1330         NG(IST)=VG(J)
1340         20 V(J)=YTMP
1350         DO 25 K=1,NG
1360             IF(K.IST)
1370                 AMIN=1.E+36
1380                 DO 27 L=IST,NG
1390                     IF(VG(L).GT.AMIN) GO TO 27
1400                     AMIN=VG(L)
1410                     L=1
1420                 27 CONTINUE
1430                 YTMP=VG(IST)
1440                 VG(IST)=VG(L)
1450                 25 VG(L)=YTMP
1460 C ELIMINATE EQUAL XG AND YG VALUES.
1470         NGH1=NG-1
1480         IZG=0
1490         DO 30 I=1,NGH1
1500             IF(NG(I).NE.NG(I+1)) GO TO 30
1510             IZG=IZG+1
1520             IZG(I)=I
1530             30 I=I+1

```

```

1170      DO CONTINUE
1170      NGX=NG
1170      IF(IIEQ.EQ.0) GO TO 100
1170      DO 40 I=1,IIEQ
1180      NGX=NGX-1
1200      IST=IEQ(I)
1210      DO 50 II=IST,NGX
1220      50 YG(II)=YG(II+1)
1230      40 CONTINUE
1240      100 IIEQ=0
1250      DO 60 I=1,NGM1
1260      IF(YG(I).NE.YG(I+1)) GO TO 60
1270      IIEQ=IIEQ+1
1280      IEQ(IIEQ)=I
1290      60 CONTINUE
1300      NGY=NG
1310      IF(IIEQ.EQ.0) GO TO 200
1320      DO 70 I=1,IIEQ
1330      NGY=NGY-1
1340      IST=IEQ(I)
1350      DO 80 II=IST,NGY
1360      80 YG(II)=YG(II+1)
1370      70 CONTINUE
1380      C MATCH CORNER POINTS WITH GRID POINTS
1390      C ICHC(10) IS GRID POINT CORRESPONDING TO CORNER POINT 10
1400      DO 90 IC=1,NC
1410      DO 90 IG=1,NGX
1420      IF(ICHC(IC).EQ.YG(IG)) ICHC(IC)=IG
1430      90 CONTINUE
1440      DO 95 JC=1,NC
1450      DO 95 JG=1,NGY
1460      IF(YG(JG).EQ.YG(JG)) JGYC(JC)=JG
1470      95 CONTINUE
1480      C IDENTIFY THE VOID AND SOLID REGIONS.
1490      C 100 REFERS TO REGION WHICH IS ABOVE AND TO THE LEFT OF POINT 100
1500      C CODE: VOID=0, SOLID=1, SOLID=2, UNASSIGNED=3.
1510      NGXP1=NGX+1
1520      NGYP1=NGY+1
1530      DO 101 I=1,NGXP1
1540      DO 101 J=1,NGYP1
1550      101 ISOLID(I,J)=3
1560      DO 999 M=1,NC
1570      N=M+1
1580      IF(M.EQ.NC) N=1
1590      IC=0
1600      IF(YG(N).GT.YG(M)) IC=1
1610      IF(ICHC(N).GT.ICHC(M)) IC=2
1620      IF(YG(N).LT.YG(M)) IC=3
1630      IF(ICHC(N).LT.ICHC(M)) IC=4
1640      DO 110 I10,120,130,140,IC
1650      110 IN=ICHC(N)
1660      JN=JGYC(N)
1670      JN=JGYC(N)
1680      JN=JN+1
1690      DO 112 J=JN,JN
1700      ISOLID(IN,J)=0
1710      112 ISOLID(IN+1,J)=1
1720      GO TO 999
1730      120 IM=ICHC(M)
1740      IM=IM+1
1750      JN=JGYC(M)
1760      IN=ICHC(M)
1770      DO 122 I=IM,IN
1780      ISOLID(I,JN)=1
1790      122 ISOLID(I,JN+1)=0
1800      GO TO 999

```

```

1810 130 IN=IGXC(N)
1820 JN=JGYC(N)
1830 JN=JN+1
1840 JN=JGYC(M)
1850 DO 122 J=JN,JM
1860 ISOLID(IN,J)=1
1870 132 ISOLID(IN+1,J)=0
1880 GO TO 999
1890 140 IN=IGXC(N)
1900 IN=IN+1
1910 JN=JGYC(N)
1920 JM=IGXC(M)
1930 DO 142 I=IN,IM
1940 ISOLID(I,JN)=0
1950 142 ISOLID(I,JN+1)=1
1960 999 CONTINUE
1970 C SET BOUNDARIES TO VOID
1980 DO 310 J=1,NGYF1
1990 ISOLID(I,J)=0
2000 310 ISOLID(NGXF1,J)=0
2010 DO 320 I=1,NGXF1
2020 ISOLID(I,1)=0
2030 320 ISOLID(I,NGYF1)=0
2040 C USE Y SWEEP TO SET ALL OTHERS
2050 DO 400 I=2,NGX
2060 DO 410 J=1,NGYF1
2070 IF (ISOLID(I,J).NE.0) GO TO 410
2080 J1=J
2090 DO 420 JJ=J1,NGYF1
2100 IF (ISOLID(I,JJ).EQ.0) GO TO 420
2110 J2=JJ-1
2120 GO TO 430
2130 420 CONTINUE
2140 WRITE(6,219)
2150 219 FORM 'X Y Z'
2160 DO 425 JJ=J1,J2
2170 425 ISOLID(I,JJ)=ISOLID(I,J1-1)
2180 410 CONTINUE
2190 400 CONTINUE
2200 C REDEFINE: ISOLID(I,J)=2 IS MATERIAL 2
2210 DO 405 I=2,NGX
2220 DO 405 J=2,NGY
2230 XMID=(XG(I)+XG(I-1))* .5
2240 YMID=(YG(J)+YG(J-1))* .5
2250 IF ((XMID.GT.XMIN1).AND.(XMID.LT.XMAX1)) GO TO 405
2260 GO TO 405
2270 405 IF ((YMID.GT.YMIN1).AND.(YMID.LT.YMAX1)) ISOLID(I,J)=2
2280 405 CONTINUE
2290 SMY=0.
2300 SMX=0.
2310 AREA1=0.
2320 DO 500 I=2,NGX
2330 DO 500 J=2,NGY
2340 ISOL=ISOLID(I,J)
2350 IF (ISOL.EQ.0) ISOL=0
2360 DA=(XG(I)-XG(I-1))*(YG(J)-YG(J-1))*ISOL
2370 AREA1=AREA1+DA
2380 X=(XG(I)+XG(I-1))* .5
2390 Y=(YG(J)+YG(J-1))* .5
2400 SMY=SMY+DA*X
2410 500 SMX=SMX+DA*Y
2420 XBAR1=SMY/AREA1
2430 YBAR1=SMX/AREA1
2440 AIX1=0.
2450 AIY1=0.
2460 AIXY1=0.

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2470      RC1=RX1+XC(1)-XBAR1
2480      DO 510 I=2,NGX
2490      DO 510 J=2,NGY
2500      ISOL=ISOLID(I,J)
2510      IF (ISOL.EQ.2) ISOL=0
2520      DX=XG(I)-XG(I-1)
2530      DY=YG(J)-YG(J-1)
2540      DA=DX*DY
2550      X=(XG(I)+XG(I-1))* .5
2560      Y=(YG(J)+YG(J-1))* .5
2570      X=X-XBAR1
2580      Y=Y-YBAR1
2590      AIX1=AIX1+(DX*DY**3/12. +DA*Y*Y)*ISOL
2600      AIY1=AIY1+(DY*DX**3/12. +DA*X*X)*ISOL
2610 510 AIXY1=AIXY1+X*Y*DA*ISOL
2620      SMY=0.
2630      SMX=0.
2640      AREA2=0.
2650      DO 501 I=2,NGX
2660      DO 501 J=2,NGY
2670      ISOL=ISOLID(I,J)
2680      IF (ISOL.EQ.1) ISOL=0
2690      IF (ISOL.EQ.2) ISOL=1
2700      DA=(XG(I)-XG(I-1))*(YG(J)-YG(J-1))*ISOL
2710      AREA2=AREA2+DA
2720      X=(XG(I)+XG(I-1))* .5
2730      Y=(YG(J)+YG(J-1))* .5
2740      SMY=SMY+DA*X
2750 501 SMX=SMX+DA*Y
2760      XBAR2=SMY/AREA2
2770      YBAR2=SMX/AREA2
2780      AIX2=0.
2790      AIY2=0.
2800      AIXY2=0.
2810      RC2=RX1+XC(1)-XBAR2
2820      DO 511 I=2,NGX
2830      DO 511 J=2,NGY
2840      ISOL=ISOLID(I,J)
2850      IF (ISOL.EQ.1) ISOL=0
2860      IF (ISOL.EQ.2) ISOL=1
2870      DX=XG(I)-XG(I-1)
2880      DY=YG(J)-YG(J-1)
2890      DA=DY*DX
2900      X=(XG(I)+XG(I-1))* .5
2910      Y=(YG(J)+YG(J-1))* .5
2920      X=X-XBAR2
2930      Y=Y-YBAR2
2940      AIX2=AIX2+(DX*DY**3/12. +DA*Y*Y)*ISOL
2950      AIY2=AIY2+(DY*DX**3/12. +DA*X*X)*ISOL
2960 511 AIXY2=AIXY2+X*Y*DA*ISOL
2970 C REFINE THE MESH
2980      JY=0.
2990      IX=0.
3000      NGXM1=NGX-1
3010      DO 540 I=1,NGXM1
3020      IGN(I)=(XG(I+1)-XG(I))/DXMAX+.999
3030      DXUSED=(XG(I+1)-XG(I))/IGN(I)
3040      IGNI=IGN(I)
3050      DO 520 II=1,IGNI
3060      IX=IX+1
3070      XGF(IX)=XG(I)+DXUSED*(II-1)
3080 520 CONTINUE
3090      IF (I.EQ. NGXM1) XGF(IX+1)=XG(I+1)
3100 540 CONTINUE
3110      NGYM1=NGY-1
3120      DO 550 I=1,NGYM1

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3130      JGN(I)=(YG(I+1)-YG(I))/DYMAX+.999
3140      DYUSED=(YG(I+1)-YG(I))/JGN(I)
3150      JGNI=JGN(I)
3160      DO 530 JJ=1,JGNI
3170      JY=JY+1
3180      YGP(JY)=YG(I)+DYUSED*(JJ-1)
3190      530 CONTINUE
3200      IF(I.EQ.NGYM1) YGP(JY+1)=YG(I+1)
3210      550 CONTINUE
3220 C SET ADDITIONAL MESH AREAS TO SOLID OR VOID.
3230      IXMAX=IX+1
3240      JYMAX=JY+1
3250      JBEG=1
3260      IBEG=1
3270      NPGXP1=IXMAX+1
3280      NPGYP1=JYMAX+1
3290 C WRITE(7,*) IXMAX,JYMAX,NPGXP1,NPGYP1,RX1,XC(1)
3300      DO 560 II=1,NPGXP1
3310      DO 560 JJ=1,NPGYP1
3320      560 IPSOLI(II,JJ)=0.
3330      DO 595 I=2,NGX
3340      JBEG=1
3350      DO 590 J=2,NGY
3360      IEND=IBEG+IGN(I-1)
3370      JEND=JBEG+JGN(J-1)
3380      IBEGP1=IBEG+1
3390      JBEGP1=JBEG+1
3400      IF(ISOLID(I,J).NE.0) GO TO 570
3410      GO TO 590
3420      570 CONTINUE
3430      DO 580 II=IBEGP1,IEND
3440      DO 580 JJ=JBEGP1,JEND
3450      580 IPSOLI(II,JJ)=1
3460      590 JBEG=JGN(J-1)+JBEG
3470      595 IBEG=IGN(I-1)+IBEG
3480 C REDEFINE IPSOLI(I,J)=2 IS THE CARBON
3490      DO 600 I=2,IXMAX
3500      DO 600 J=2,JYMAX
3510      XMID=(XGP(I)+XGP(I-1))* .5
3520      YMID=(YGP(J)+YGP(J-1))* .5
3530      IF((XMID.GT.XMIN1).AND.(XMID.LT.XMAX1)) GO TO 601
3540      GO TO 600
3550      601 IF((YMID.GT.YMIN1).AND.(YMID.LT.YMAX1)) IPSOLI(I,J)=2
3560      600 CONTINUE
3570 C CHECK TO SEE IF TOTAL AREA IS THE SAME.
3580      AREA=0.
3590      DO 700 I=2,IXMAX
3600      DO 700 J=2,JYMAX
3610      IPSOL=IPSOLI(I,J)
3620      IF(IPSOL.EQ.2) IPSOL=1
3630      DA=(XGP(I)-XGP(I-1))*(YGP(J)-YGP(J-1))*IPSOL
3640      AREA=AREA+DA
3650      700 CONTINUE
3660      DO 702 I=1,IXMAX
3670      702 XGP(I)=RX1-XGP(I)
3680      DIFF=AREA1+AREA2-AREA
3690 C
3700      WRITE(6,1)
3710      1 FORMAT(//,5X,'INPUT GRID LOCATIONS',/,5X,'XC:')
3720      WRITE(6,2) XC
3730      2 FORMAT(9X,9F7.4)
3740      WRITE(6,3)
3750      3 FORMAT(5X,'YC:')
3760      WRITE(6,2) YC
3770      WRITE(6,4)
3780      4 FORMAT(//,5X,'ADJUSTED GRID LOCATIONS',/,5X,'XG:')

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3730      WRITE(6,2) (XG(I), I=1, NGX)
3800      WRITE(6,5)
3810      5 FORMAT(5X, 'YG:')
3820      WRITE(6,2) (YG(I), I=1, NGY)
3830      WRITE(6,6)
3840      6 FORMAT(/, 5X, 'SECTION PROPERTIES MATRIX', /)
3850      DO 7 I=1, NGXP1
3860      7 WRITE(6,8) (ISOLID(I, J), J=1, NGYP1)
3870      8 FORMAT(2X, 40I2)
3880      WRITE(6,9) AREA1, XBAR1, YBAR1, RC1, AIX1, AIY1, AIXY1
3890      9 FORMAT(/, 5X, 'AREA1=', E11. 4, 3X, 'XBAR1=', E11. 4, 3X, 'YBAR1=',
3900      *E11. 4, 3X, 'RC1=', E11. 4, /, 5X, 'IX1=', E11. 4, 3X, 'IY1=', E11. 4,
3910      *3X, 'IXY1=', E11. 4)
3920      WRITE(6,17) AREA2, XBAR2, YBAR2, RC2, AIX2, AIY2, AIXY2
3930      17 FORMAT(/, 5X, 'AREA2=', E11. 4, 3X, 'XBAR2=', E11. 4, 3X, 'YBAR2=',
3940      *E11. 4, 3X, 'RC2=', E11. 4, /, 5X, 'IX2=', E11. 4, 3X, 'IY2=', E11. 4,
3950      *23X, 'IXY2=', E11. 4)
3960      WRITE(6,11)
3970      11 FORMAT(/, 5X, 'FINER MESH ADJUSTMENT', /, 5X, 'XGP:')
3980      WRITE(6,2) (XGP(I), I=1, IXMAX)
3990      WRITE(7,2) (XGP(I), I=1, IXMAX)
4000      WRITE(6,12)
4010      12 FORMAT(5X, 'YGP:')
4020      WRITE(6,2) (YGP(I), I=1, JYMAX)
4030      WRITE(7,2) (YGP(I), I=1, JYMAX)
4040      IF(NPGXP1.GT.40) GO TO 705
4050      WRITE(6,13)
4060      13 FORMAT(/, 5X, 'REFINED MESH', /)
4070      DO 14 I=1, NPGXP1
4080      WRITE(7,8) (IPSOLI(I, J), J=1, NPGYP1)
4090      14 WRITE(6,8) (IPSOLI(I, J), J=1, NPGYP1)
4100      705 CONTINUE
4110      WRITE(6,15) DIFF
4120      15 FORMAT(/, 5X, 'DIFFERENCE IN AREAS=', E11. 4)
4130      ERR=0.01
4140      IMAXM1=IXMAX-1
4150      JMAXM1=JYMAX-1
4160      C HEAT TRANSFER EQUATION SETUP
4170      C FIND THE NUMBER OF ZEROS
4180      DO 710 I=1, IXMAX
4190      DO 710 J=1, JYMAX
4200      QS=0.
4210      H=0.
4220      IS(2)=IPSOLI(I, J)
4230      IS(1)=IPSOLI(I, J+1)
4240      IS(3)=IPSOLI(I+1, J)
4250      IS(4)=IPSOLI(I+1, J+1)
4260      NZGO=0
4270      DO 800 K=1, 4
4280      IF(IS(K).EQ.0) NZGO=NZGO+1
4290      800 CONTINUE
4300      C FIND WHICH ORIGINAL BOUNDARY YOU ARE ON
4310      IC=0
4320      DO 810 K=1, NC
4330      KP1=K+1
4340      IF(KP1.GT.NC) KP1=1
4350      IF(XC(K).EQ.XGP(I)) GO TO 812
4360      IF(YC(K).EQ.YGP(J)) GO TO 814
4370      GO TO 810
4380      812 IF(YC(K).EQ.YGP(J)) GO TO 820
4390      IF((YGP(J).GT.YC(K)).AND.(YGP(J).LT.YC(KP1))) GO TO 825
4400      IF((YGP(J).LT.YC(K)).AND.(YGP(J).GT.YC(KP1))) GO TO 825
4410      GO TO 810
4420      814 IF((XGP(I).GT.XC(K)).AND.(XGP(I).LT.XC(KP1))) GO TO 830
4430      IF((XGP(I).LT.XC(K)).AND.(XGP(I).GT.XC(KP1))) GO TO 830
4440      C NOT A BOUNDARY POINT

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4450      810 CONTINUE
4460      GO TO 900
4470 C THIS IS A CORNER POINT
4480      820 IC=K
4490      GO TO 900
4500 C POINT IS ON X=CONST LINE BETWEEN ICX AND ICX+1
4510      825 IC=K
4520      GO TO 900
4530 C POINT IS ON Y=CONST LINE BETWEEN ICY AND ICY+1
4540      830 IC=K
4550 C WATCH OUT - EITHER WAY ALGEBRAICALLY
4560 C CALCULATE CONDUCTIVE COEFFICIENTS
4570 C NOTE - SOME QUANTITIES BELOW ARE UNDEFINED
4580      900 ICP1=IC+1
4590      IF(ICP1.GT.NC) ICP1=1
4600      DYJP1=YGF(J+1)-YGF(J)
4610      DYJM1=YGF(J)-YGF(J-1)
4620      DXIP1=XGF(I+1)-XGF(I)
4630      DXIM1=XGF(I)-XGF(I-1)
4640      IF(DYJP1.EQ.0.) DYJP1=2.
4650      IF(DYJM1.EQ.0.) DYJM1=2.
4660      IF(DXIP1.EQ.0.) DXIP1=2.
4670      IF(DXIM1.EQ.0.) DXIM1=2.
4680 C NOTE - COND(0) IS UNDEFINED
4690      COND1=COND(IS(1))
4700      COND2=COND(IS(2))
4710      COND3=COND(IS(3))
4720      COND4=COND(IS(4))
4730      IF(IS(1).EQ.0) COND1=0.
4740      IF(IS(2).EQ.0) COND2=0.
4750      IF(IS(3).EQ.0) COND3=0.
4760      IF(IS(4).EQ.0) COND4=0.
4770 C THESE ARE ZERO FOR COND ZERO
4780      KIM1=0.5*(COND1*DYJP1+COND2*DYJM1)*2.*PIE*(RGP(I)+.5*DXIM1)/DXIM1
4790      KIP1=0.5*(COND4*DYJP1+COND3*DYJM1)*2.*PIE*(RGP(I)-.5*DXIP1)/DXIP1
4800      KJM1=COND2*.5*DXIM1*2.*PIE*(RGP(I)+.25*DXIM1)
4810      KJM1=(KJM1+COND3*.5*DXIP1*2.*PIE*(RGP(I)-.25*DXIP1))/DYJM1
4820      KJP1=COND1*.5*DXIM1*2.*PIE*(RGP(I)+.25*DXIM1)
4830      KJP1=(KJP1+COND4*.5*DXIP1*2.*PIE*(RGP(I)-DXIP1*.25))/DYJP1
4840 C GO TO THE SELECTED CASE
4850      NZGO=NZGO+1
4860      GO TO (1000,1100,1200,1300,1400),NZGO
4870 C CLL SOLID - CASE 14
4880 C WILL ASSUME HEAT INPUT ON Y=CONST LINE IF MATLS ARE DIFFERENT
4890 C ON EACH SIDE - IE - SLIDING INTERFACE
4900 C Q MUST BE HEAT GENERATED PER AREA
4910      1000 Q5=0.
4920      IF(IS(1).EQ.IS(2)) GO TO 1010
4930      AREA=2.*PIE*(RGP(I)+.25*DXIM1)*DXIM1*.5+2.*PIE*(RGP(I)-.25*DXIP1)*
4940      *DXIP1*.5
4950      Q5=Q*AREA
4960      1010 AJ=KIP1+KIM1+KJP1+KJM1
4970      AIP1(I,J)=KIP1/AJ
4980      AIM1(I,J)=KIM1/AJ
4990      AJP1(I,J)=KJP1/AJ
5000      AJM1(I,J)=KJM1/AJ
5010      BIJ(I,J)=Q5/AJ
5020      GO TO 710
5030 C ONE ZERO
5040      1100 IF(IS(1).EQ.0) GO TO 1110
5050      IF(IS(2).EQ.0) GO TO 1120
5060      IF(IS(3).EQ.0) GO TO 1130
5070      IF(IS(4).EQ.0) GO TO 1140
5080      WRITE(6,1111)
5090      1111 FORMAT(' WHOOPS')
5100 C CASE 10

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5110 1110 QS=0.
5120 IF (IS(3).EQ. IS(4)) GO TO 1115
5130 AREA=2. *PIE*(RGP(I)-. 25*DXIP1)*. 5*DXIP1
5140 QS=Q+AREA
5150 1115 H=2. *PIE*RGP(I)*. 5*DYJP1*HCP(IC)
5160 H=H+2. *PIE*(RGP(I)+DXIM1*. 25)*. 5*DXIM1*HCP(IC)
5170 GO TO 1150
5180 C CASE 9
5190 1120 QS=0.
5200 IF (IS(3).EQ. IS(4)) GO TO 1125
5210 AREA=2. *PIE*(RGP(I)-. 25*DXIP1)*. 5*DXIP1
5220 QS=Q+AREA
5230 1125 H=2. *PIE*RGP(I)*. 5*DYJM1*HCP(IC)
5240 H=H+2. *PIE*(RGP(I)+DXIM1*. 25)*. 5*DXIM1*HCP(IC)
5250 GO TO 1150
5260 C CASE 12
5270 1130 QS=0.
5280 IF (IS(2).EQ. IS(1)) GO TO 1135
5290 AREA=2. *PIE*(RGP(I)+. 25*DXIM1)*. 5*DXIM1
5300 QS=Q+AREA
5310 1135 H=2. *PIE*RGP(I)*. 5*DYIM1*HCP(IC)
5320 H=H+1. *PIE*(RGP(I)-. 25*DXIP1) *. 5*DXIP1*HCP(IC)
5330 GO TO 1150
5340 C CASE 11
5350 1140 QS=0.
5360 IF (IS(2).EQ. IS(1)) GO TO 1145
5370 AREA=2. *PIE*(RGP(I)+. 25*DXIM1)*DXIM1*. 5
5380 QS=Q+AREA
5390 1145 H=2. *PIE*RGP(I)*. 5*DYJP1*HCP(IC)
5400 H=H+2. *PIE*(RGP(I)-. 25*DXIP1)*. 5*DXIP1*HCP(IC)
5410 1150 AJ=KIP1+KIM1+KJP1+KJM1+H
5420 AIP1(I,J)=KIP1/AJ
5430 AIM1(I,J)=KIM1/AJ
5440 AJP1(I,J)=KJP1/AJ
5450 AJM1(I,J)=KJM1/AJ
5460 BIJ(I,J)=(QS+H*TINF)/AJ
5470 GO TO 710
5480 C TWO ZEROS
5490 1200 DO 1201 II=1,4
5500 IF1=II+1
5510 IF (IP1.EQ. 5) IF1=1
5520 IF ((IS(II)+IS(IP1)).NE. 0) NGO=II
5530 1201 CONTINUE
5540 GO TO (1230,1240,1210,1220), NGO
5550 C CASE 1
5560 1210 QS=0.
5570 IF (IS(3).NE. IS(4)) QS=2. *PIE*(RGP(I)-. 25*DXIP1)*. 5*DXIP1+Q
5580 YM=YGP(J)-DYJM1*. 5
5590 YP=YGP(J)+DYJP1*. 5
5600 IF (YCHG(IC).NE. 0.) GO TO 1211
5610 H=2. *PIE*RGP(I)*(YP-YM)*HCP(IC)
5620 GO TO 1250
5630 1211 FRAC=(YCHG(IC)-YM)/(YP-YM)
5640 IF (FRAC.LT. 0.) FRAC=0.
5650 IF (FRAC.GT. 1.) FRAC=1.
5660 H=(FRAC*HCP(IC)+(1.-FRAC)*HCP(ICP1))*2. *PIE*RGP(I)*(YP-YM)
5670 GO TO 1250
5680 C CASE 2
5690 1220 XP=XGP(I)+DXIP1*. 5
5700 XM=XGP(I)-DXIM1*. 5
5710 RAV=RX1-(XP+XM)*. 5
5720 IF (XCHG(IC).NE. 0.) GO TO 1221
5730 H=2. *PIE*RAV*(XP-XM)*HCP(IC)
5740 GO TO 1250
5750 1221 FRAC=(XCHG(IC)-XM)/(XP-XM)
5760 IF (FRAC.LT. 0.) FRAC=0.

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5770      IF(FRAC.GT.1.) FRAC=1.
5780      H=(FRAC*HCM(ICP1)+(1.-FRAC)*HCP(IC))*2.*PIE*RAY*(XP-XM)
5790      GO TO 1250
5800 C CASE 3
5810      1230 QS=0.
5820      IF(IS(2).NE.IS(1)) QS=2.*PIE*(RGP(I)+.25*DXIM1)*.5*DXIM1*Q
5830      YM=YGP(J)-DYJM1*.5
5840      YP=YGP(J)+DYJP1*.5
5850      IF(YCHG(IC).NE.0.) GO TO 1231
5860      H=2.*PIE*RGP(I)*(YP-YM)*HCP(IC)
5870      GO TO 1250
5880      1231 FRAC=(YCHG(IC)-YM)/(YP-YM)
5890      IF(FRAC.LT.0.) FRAC=0.
5900      IF(FRAC.GT.1.) FRAC=1.
5910      H=(FRAC*HCM(ICP1)+(1.-FRAC)*HCP(IC))*2.*PIE*RGP(I)*(YP-YM)
5920      GO TO 1250
5930 C CASE 4
5940      1240 XP=XGP(I)+DXIP1*.5
5950      XM=XGP(I)-DXIM1*.5
5960      RAV=RX1-(XP+XM)*.5
5970      IF(XCHG(IC).NE.0.) GO TO 1241
5980      H=2.*PIE*RAV*(XP-XM)*HCP(IC)
5990      GO TO 1250
6000      1241 FRAC=(XCHG(IC)-XM)/(XP-XM)
6010      IF(FRAC.LT.0.) FRAC=0.
6020      IF(FRAC.GT.1.) FRAC=1.
6030      H=(FRAC*HCP(IC)+(1.-FRAC)*HCM(ICP1))*2.*PIE*RAV*(XP-XM)
6040      1250 AJ=KIP1+KIM1+KJP1+KJM1+H
6050      AIP1(I,J)=KIP1/AJ
6060      AIM1(I,J)=KIM1/AJ
6070      AJP1(I,J)=KJP1/AJ
6080      AJM1(I,J)=KJM1/AJ
6090      BIJ(I,J)=(QS+H*TINF)/AJ
6100      GO TO 710
6110 C THREE ZEROS
6120      1300 DO 1301 II=1,4
6130      IF(IS(II).NE.0) NGO=II
6140      1301 CONTINUE
6150      GO TO (1310,1320,1330,1340),NGO
6160 C CASE 5
6170      1310 RAV=RGP(I)+DXIM1*.25
6180      H=2.*PIE*(RAV*DXIM1*.5*HCP(IC)+RGP(I)*DYJP1*.5*HCM(IC))
6190      GO TO 1350
6200 C CASE 6
6210      1320 RAV=RGP(I)+DXIM1*.25
6220      H=2.*PIE*(RAV*DXIM1*.5*HCM(IC)+RGP(I)*DYJM1*.5*HCP(IC))
6230      GO TO 1350
6240 C CASE 7
6250      1330 RAV=RGP(I)-DXIP1*.25
6260      H=2.*PIE*(RAV*DXIP1*.5*HCP(IC)+RGP(I)*DYJM1*.5*HCM(IC))
6270      GO TO 1350
6280 C CASE 8
6290      1340 RAV=RGP(I)-DXIP1*.25
6300      H=2.*PIE*(RAV*DXIP1*.5*HCM(IC)+RGP(I)*DYJP1*.5*HCP(IC))
6310      1350 AJ=KIP1+KIM1+KJP1+KJM1+H
6320      AIP1(I,J)=KIP1/AJ
6330      AIM1(I,J)=KIM1/AJ
6340      AJP1(I,J)=KJP1/AJ
6350      AJM1(I,J)=KJM1/AJ
6360      BIJ(I,J)=(QS+H*TINF)/AJ
6370      GO TO 710
6380 C FOUR ZEROS - VOID - CASE 13
6390      1400 AIP1(I,J)=0.
6400      AIM1(I,J)=0.
6410      AJP1(I,J)=0.
6420      AJM1(I,J)=0.

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6430      BIJ(I,J)=0.
6440 710      CONTINUE
6450 1947      FORMAT(1H , 2I3, 6E12. 4)
6460      DO 720 I=1, IXMAX
6470      DO 720 J=1, JYMAX
6480 720      T(I,J)=TINF
6490      OMEG=1. 7
6500      OMO=1. -OMEG
6510      DO 740 IT=1, 100
6520      AMAX=0. 0
6530      DO 730 I=1, IXMAX
6540      DO 730 J=1, JYMAX
6550      TOLD=T(I,J)
6560      TNEW=AIM1(I,J)*T(I-1,J)+AIP1(I,J)*T(I+1,J)+AJM1(I,J)*T(I,J-1)
6570      TNEW=TNEW+AJF1(I,J)*T(I,J+1)+BIJ(I,J)
6580      T(I,J)=TNEW+OMEG+OMO*TOLD
6590      VAL=ABS(T(I,J)-TOLD)
6600      IF(VAL.GT.AMAX) AMAX=VAL
6610 730      CONTINUE
6620      IF(AMAX.LT.ERR) GO TO 750
6630 740      CONTINUE
6640      WRITE(6,745) IT
6650 745      FORMAT(5X, 'EXCEEDS ITERATIONS. IT=', I4)
6660 750      CONTINUE
6670      DO 610 J=1, JYMAX
6680      DO 611 I=1, IXMAX
6690      II=IXMAX+1-I
6700 611      TTT(II)=T(I,J)
6710 610      WRITE(6,612) (TTT(II), II=1, IXMAX)
6720 612      FORMAT(1H , 12F5. 0)
6730 C EXACT SECTION PROPERTIES AND THERMAL ROTATION
6740      JX1=0.
6750      JX2=0.
6760      JY1=0.
6770      JY2=0.
6780      JXY1=0.
6790      JXY2=0.
6800      YT1=0.
6810      YT2=0.
6820      T1=0.
6830      T2=0.
6840      DO 2000 I=1, IXMAX
6850      DO 2000 J=2, JYMAX
6860      S1=0.
6870      S2=0.
6880      IF(IPSOLI(I,J).EQ.1) S1=1.
6890      IF(IPSOLI(I,J).EQ.2) S2=1.
6900      DA=(XGP(I)-XGP(I-1))*(YGP(J)-YGP(J-1))
6910      YY=(YGP(J)+YGP(J-1))* .5
6920      XX=(XGP(I)+XGP(I-1))* .5
6930      XX1=XX-XBAR1
6940      YY1=YY-YBAR1
6950      XX2=XX-XBAR2
6960      YY2=YY-YBAR2
6970      TT=(T(I,J)+T(I-1,J)+T(I,J-1)+T(I-1,J-1))* .25
6980 C DEFLECTION IS BASED ON TEMP RISE RELATIVE TO TINF
6990      TT=TT-TINF
7000      DEN1=1. -XX1/RC1
7010      DEN2=1. -XX2/RC2
7020      JX1=JX1+YY1*YY1*DA*S1/DEN1
7030      JY1=JY1+XX1*XX1*DA*S1/DEN1
7040      JXY1=JXY1+XX1*YY1*DA*S1/DEN1
7050      T1=T1+TT*DA*S1
7060      YT1=YT1+TT*YY1*DA*S1
7070      JX2=JX2+YY2*YY2*DA*S2/DEN2
7080      JY2=JY2+XX2*XX2*DA*S2/DEN2

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7090      JXY2=JXY2+XX2*YY2*DA*S2/DEN2
7100      T2=T2+TT*DA*S2
7110      YT2=YT2+TT*YY2*DA*S2
7120 2000 CONTINUE
7130      Q1=AREA1*RC1*RC1+JY1
7140      PHI1=RC1*(Q1*YT1-RC1*JXY1*T1)/(JX1*Q1-JXY1**2)
7145      PHI1=PHI1*ALPHA1
7150      Q2=AREA2*RC2*RC2+JY2
7160      PHI2=RC2*(Q2*YT2-RC2*JXY2*T2)/(JX2*Q2-JXY2**2)
7165      PHI2=PHI2*ALPHA2
7166      PHI=PHI1-PHI2
7170      WRITE(6,2005) JX1,JY1,JXY1,PHI1
7180      WRITE(6,2010) JX2,JY2,JXY2,PHI2
7190 2005 FORMAT(1H,'JX1=',E12.4,'JY1=',E12.4,'JXY1=',E12.4,'PHI1=',E12.
7200      *4)
7210 2010 FORMAT(1H,'JX2=',E12.4,'JY2=',E12.4,'JXY2=',E12.4,'PHI2=',E12.
7220      *4)
7222      AKTHRM=PHI/(QTOT/(2.*PIE*RFAVG))
7225      WRITE(6,2015) PHI,AKTHRM
7226 2015 FORMAT('PHI=',E12.4,'K THERMAL=',E12.4)
7230      STOP
7240      END

```

HTSEAL 07/17/84 13:03:25
QTOT= 0.2797E+04

INPUT GRID LOCATIONS

XC: 0.0054 0.0054 0. 0. 0.0067 0.0067 0.0026 0.0026 0.0125
0.0125 0.0114 0.0114 0. 0. 0. 0. 0. 0.
0. 0. 0. 0.
YC: 0. 0.0152 0.0152 0.0286 0.0286 0.0318 0.0318 0.0517 0.0517
0.0318 0.0318 0. 0. 0. 0. 0. 0.
0. 0. 0. 0.

ADJUSTED GRID LOCATIONS

XG: 0. 0.0026 0.0054 0.0067 0.0114 0.0125
VG: 0. 0.0152 0.0286 0.0318 0.0517

SECTION PROPERTIES MATRIX

0 0 0 0 0 0
0 0 1 0 0 0
0 0 1 0 2 0
0 1 1 0 2 0
0 1 1 1 2 0
0 0 0 0 2 0
0 0 0 0 0 0

AREA1= 0.1035E-03 XBAR1= 0.5000E-01 YBAR1= 0.1700E-01 RC1= 0.5010E-0
IX1= 0.1828E-07 IV1= 0.2461E-08 IXV1= 0.1765E-08

AREA2= 0.1986E-03 XBAR2= 0.7525E-02 YBAR2= 0.4173E-01 RC2= 0.5757E-0
IX2= 0.6594E-08 IV2= 0.1639E-08 IXV2= 0.1334E-19

FINER MESH ADJUSTMENT

XGP: 0. 0.0013 0.0026 0.0040 0.0054 0.0067 0.0079 0.0090 0.0102
0.0114 0.0125
YGP: 0. 0.0014 0.0028 0.0042 0.0055 0.0069 0.0083 0.0097 0.0111
0.0125 0.0139 0.0152 0.0167 0.0182 0.0197 0.0212 0.0227 0.0241
0.0256 0.0271 0.0286 0.0296 0.0307 0.0318 0.0332 0.0346 0.0360
0.0375 0.0389 0.0403 0.0417 0.0432 0.0446 0.0460 0.0474 0.0489
0.0503 0.0517

REFINED MESH

0
0 0 0 0 0 0 0 0 0 0 0 0 1 1 1 1 1 1 1 1 0 0 0 0 0 0 0 0 0 0 0 0
0 0 0 0 0 0 0 0 0 0 0 0 1 1 1 1 1 1 1 1 0 0 0 0 0 0 0 0 0 0 0 0
0 0 0 0 0 0 0 0 0 0 0 0 1 1 1 1 1 1 1 1 0 0 0 2 2 2 2 2 2 2 2 2
0 0 0 0 0 0 0 0 0 0 0 0 1 1 1 1 1 1 1 1 0 0 0 2 2 2 2 2 2 2 2 2
0 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 0 0 0 2 2 2 2 2 2 2 2 2
0 1 2 2 2 2 2 2 2 2 2 2
0 1 2 2 2 2 2 2 2 2 2
0 1 2 2 2 2 2 2 2 2 2
0 1 2 2 2 2 2 2 2 2 2
0 2 2 2 2 2 2 2 2 2
0 0

[illegible]

F19